

CONTACT INFORMATION

Mining Records Curator Arizona Geological Survey 1520 West Adams St. Phoenix, AZ 85007 602-771-1601 http://www.azgs.az.gov inquiries@azgs.az.gov

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Regarding an exploration license and purchase option, the owners would require the three drill holes mentioned earlier, aligned to intercept the Wasp ore shoot at depth at it's known point of highest assay. These should be cored. Calculations for this alignment will be agreed upon prior to any license. After these three boreholes, the developer may drill anywhere he wishes.

We require the results of the developer's assay reports as they are developed, and the right to sample drill cuttings and split drill cores. The capacity of the Wasp lab will not handle the work of two operators. Unless otherwise agreed, we retain sole control of the Wasp laboratory until any sale is funded and closed. We also require evidence of the operator's prior experience and financial ability to develop an underground mine in the probable event that it appears warranted by the drilling program.

The Old Wasp Mine is offered for sale for U.S.\$2,200,000. This consists of the 9.51 acres including the mine, machinery, mill, shop, lab, office, owner's residence with all real property improvements. Alternatively, it is available for U.S.\$2,060,000 including the mineral estate to the entire 9.51 acres, but with us retaining the surface rights to only the 1.25 acres which includes our home, shop, lab and office, with the right of ingress and egress. This 1¼-acres is across the highway from, and would not interfere with, the mining or milling operation. The mineral estate underlying this 1¼-acres would be conveyed to the buyer, who must warrant us against damage from blasting or subsidence.

Residential land in the area, with utilities available, runs from \$30,000 to \$50,000 per acre, some as high as \$100,000 per acre. This can be verified by contacting any area real estate broker.

The owner's residence on 1 1/4 acres was FHA appraised five years ago at \$92,000. and is currently rated for insurance purposes at \$140,000, which is the price reduction stipulated above in the event we retain surface title to our home. With the mill, residence, laboratory and outbuildings included, just the surface estate of the claim ought to be worth \$750,000.At a price of \$2.2 million, this is attributing only \$1.45 million to the mineral estate, as the buyer would still have the remaining value of the surface estate after the mine is depleted, which value should by then be substantially appreciated. If drilling indicates there actually is at least \$13.5 million in that one Wasp ore shoot, then a price of \$1.45 million attributed to the entire mineral estate seems reasonable.

Some speculative value must also be attributed to the fact that in 1893-1895, within pistol-shot distance, \$34 million (at today's gold prices) was taken out of two ore shoots on the adjacent claim, on a continuation of this same geological structure.

If a drilling program indicates the property is worth less than we now presume, we may consider a downward adjustment in the price, however such adjustment must be solely our prerogative. We will not index the price to drilling results in advance, as this excessively transfers control of the price to the buyer. If the developer abandons the project after the drilling program, he retains no residual earned equity in the property.

We are not soliciting lease offers, having already sustained losses of \$143,000 from the infidelity of prior lessees. In any lease event, the existing Wasp ore shoot may not be additionally mined by surface methods without our consent. Due to the history of prior operators taking the "easy pickings" and walking, the granting of such consent is questionable. Neither will we accept any investment money nor sell any fractional interests.

If a buyer/developer has other ideas, we will listen, but may not depart far from the outline offered here. Meanwhile, until someone makes a proposal, we will continue with exploration, mining and milling, during which the price and terms may be adjusted either way depending on the results of our work.

If the buyer requires the entire property, we will need a 6-month leaseback on the owner's residence to permit our orderly departure from the premises.

Clay Worst owner

THE OLD WASP MINE

FOR STUDY FURPOSES ONLY 4-25-1999

The preceding report on the Mammoth Mine contains important information that applies directly to the adjacent Old Wasp. To avoid repetition here, the reader will be referred to parts of the Mammoth Mine report that are applicable to the Wasp. The inconvenience to the reader is regrettable, but to anyone reading this entire report, an unnecessary repetition would be equally annoying.

The Old Wasp Mine is offered for sale for U.S.\$2,500,000. $H_{2,1}QQ$

The Old Wasp claim adjoins the Mammoth claim on the south, along a continuation of the Mammoth Fault hanging wall contact. While the Mammoth claim has some three miles of underground workings sunk to the 1,030 ft. level by 1925, the Wasp remained essentially a virgin property until 1983.

The Wasp is a 20.64 acre claim, patented fee simple absolute, with a quiet and undisputed title since 1893. The north half, upon which all development exists, is owned by Clay Worst and his wife, who live in the owner's residence at the mine. The south half is undeveloped, and is owned by associates have of Worst. The claim is crossed by hard-surfaced State Hwy. 88. A 3-phase electric distribution line and a 6-inch municipal water main are on the property, and a 45 kV substation is a mile distant.

The highest and best use of the Wasp would be if operated in concert with the Mammoth Mine by a major mining company. However, since the Wasp has a perfect title, a known ore body and the only currently operational mill at Goldfield, it is still a viable independent, though smaller, operation.

The preceding report on the Mammoth Mine (page 4, paragraph 11) references the work of Kennedy ⁽⁷⁾ in 1910 and Bedford ⁽¹²⁾ in 1923. If Kennedy and Bedford were correct, and the ore emplacement on the Mammoth was being fed from the south, then it was being fed from beneath the Wasp. Their counsel was undoubtedly what prompted George U. Young in 1925 to drift south at the 1,000 ft. level from the No. 7 Shaft (The Main Shaft) of the Mammoth Mine along the hanging wall contact of the Mammoth Fault, and intercept the Wasp ore shoot at the 1,000 ft. level.

That this is the same ore shoot exposed in the bottom of the present Wasp open pit is well established, for these reasons:

1) The surface exposure of the Mammoth Fault contact from the Mammoth Mine southward to the Wasp was clearly in evidence when the properties were mapped in 1916 ⁽⁷⁷⁾.

2) If the fault contact in the bottom of the Wasp open pit is protracted downward on the average westward dip of the Mammoth Fault, the calculated intercept at the 1,000 ft. level is within 5 ft. laterally of the actual location as platted on the Mammoth worksheets ⁽⁸¹⁾.

3) As the values in the Wasp pit are followed southward, they increase to a peak, and then decrease beyond that point. The same occurred at the 1,000 ft. level. If the point of highest assay in the Wasp pit is protracted downward on the dip of the vein, 90 degrees to the strike, it intercepts the point of highest assay at the 1,000 ft. level ⁽⁸¹⁾.

4) Whenever extremely high grade ore is encountered in the Wasp open pit, it is associated with malachite (copper carbonate) and galena (lead sulphide). The assay records from the Wasp shoot at the 1,000 ft. level indicate up to 3% copper and 18% lead ⁽⁸⁰⁾.

The south drift at the 1,000 ft. level in the Mammoth Mine is of great value in evaluating the Old Wasp ore shoot. The following data can be verified by the Mammoth plats of the 1,000 ft. level dated 8/20/1925 ⁽⁸⁰⁾ and 11/5/1925 ⁽⁸¹⁾.

At station 1026, the drift entered the Wasp claim. Initial values were low grade, but the continued extension of the drift was probably motivated by assays of 19.3 oz. and 21.1 oz. gold per ton, found in this drift 57 and 69 feet prior to entering the Wasp.

77 feet onto the Wasp, they encountered the beginning of another 75 feet of drift which averaged 0.70 oz. gold per ton. The central 30 feet of this 75 foot section averaged 1.12 oz. gold per ton. The 36 foot remainder of this drift averaged 0.51 oz. gold per ton. There is also a notation of 35 lineal feet of "black ore 12 in. wide" that averaged 2.52 ounces. Individual samples of "black ore" at the 1,000 ft. level assayed up to 4.94 ounces of gold per ton.

This comprises 111 feet of drift averaging 0.634 ounces of gold per ton. These were daily car samples, taken as the drift advanced, perhaps as representative samples as could be obtained.

No such "black ore" has been encountered in the Wasp open pit to date. However, in the entrance ramp to the pit, near the south end of the mined ore shoot, the operators discovered a filled-in shaft. The patent survey plat of the Wasp indicates all the shallow prospect holes, but this shaft is not shown, nor is there any known record of it.

This is a 4 x 6 foot shaft, cribbed solid with modern mill-sawed timbers, but with short lengths of hand chopped ironwood timbers laid just outside the modern timbers. None of the ironwood timbers are much over three feet long. It appears the early Anglos found an old timbered Mexican shaft, which they reopened and retimbered. The shaft is still evident at the 40 foot level on the entrance ramp. It seeps groundwater.

The shaft is filled with black material, which has not been assayed. The shaft is located above the south end of the vein of "black ore" disclosed at the 1,000 ft. level. The purpose or meaning of this shaft is unknown. No attempt to clean it out has been made, as it is in the center of the ramped roadway into the open pit, and opening it would block access to the pit.

Of significance is that a 12-inch width on the face of the completed drift at the 1,000 ft. level still averaged 3.60 ounces of gold per ton ⁽⁸¹⁾. Obviously, the drift was not discontinued due to a lack of values; it was discontinued due to the death of the operator. Accordingly, the actual southerly extent of the Wasp ore shoot at the 1,000 ft. level, and its gold values, are totally unknown.

Gold values in the Wasp open pit are not as clearly defined. No assay records of the upper 45 feet survived the original lessee's office fire.

At the 53 foot level in the open pit, assays exist only for the south end of the ore shoot, where an average width of 5.25 ft. averaged 2.20 oz. gold per ton for 30 lineal feet. At the north end of this 30 feet, an 8.2 ft. width averaged 3.48 oz. gold per ton, of which the 3.9 ft. next to the fault contact averaged 7.68 oz. gold per ton. No meaningful assay-records exist north of this point.

Usually, the highest values lie in the first two feet of granite next to the hanging wall contact. As an experiment, a 40-ft. hole was drilled in the bottom of the pit, starting one foot out into the hanging wall from the fault contact. The drill rig was "eyeball" inclined to attempt following this two-foot high-grade width downward. Samples were taken at five-foot intervals.

These samples assayed 8.7, 4.2, 3.65, 2.15, 1.90, 1.05, 0.90 and 0.75 ounces of gold per ton. This hole averaged only 2.91 ounces. It proves nothing, as it cannot be known whether the values in this two-foot width actually decreased with depth, or whether the drill hole drifted out of its estimated one-foot width tolerance. There would seem a remote chance of drilling 40-feet with an air-track drill and staying within a one-foot tolerance, with only an estimated drill inclination on an unproven dip. The only assurance was that the upper five feet averaged 8.7 ounces of gold per ton.

To interpolate values between the present 60-ft. level in the open pit and the drift at the 1,000-ft. level involves conjecture and guesswork. The point of highest value in the open pit at the 53 foot level was 3.48 ounces 8.2 ft. wide. Directly down the dip of the vein, 90 degrees to the strike, you intercept the point of highest value at the 1,000 ft. level, 1.60 ounces of gold. Accordingly, the values at this point at the 1,000 ft. level are 54% less than directly above this point in the open pit.

The most complete assay records are those at the 1,000 ft. level, where 111 feet of drift averaged 0.634 ounces. If you split the 54% variation in half for an average, a 17% increase in value in the portion of the vein overlying the 1,000 ft. level would be 0.805 ounces of gold per ton.

An ore shoot 111 feet long by 8 feet wide by 940 feet deep contains about 38,400 tons of material. At an average assay of 0.805 ounces, there would be 30,912 ounces of gold, at \$289.50 an ounce, worth \$8,949,000.

However, the south terminus of the 1,000 ft. drift still assayed 3.60 ounces 12-inches wide, so the values should not be expected to end at that point. Also, having proved only that values persist to the 1,000 ft. level, there is no reason to assume they end at that depth. With only a probable increase of 10% in length and 10% in depth, both very reasonable presumptions, the shoot would contain about \$11 million in gold values.

The foregoing arithmetic is really only conjecture, but there is no better existing data. Only a drilling program will delineate and evaluate the existing ore shoot. The remainder of the unexplored Mammoth Fault contact across the Wasp may reveal additional ore bodies.

The Wasp ore shoot ought to be drilled. Three slant bore-holes, calculated to cut the Wasp shoot at the 250 ft., 500 ft., and 750 ft. levels, would give a general idea of what values lie throughout the shoot. The outcome of that drilling may encourage additional drilling to further delineate the ore body.

The 250 ft. intercept would require 320 ft. of hole; the 500 ft. intercept would require 640 ft. of hole; the 750 ft. intercept would require 865 ft. of hole, a total of 1,825 feet of borehole; see the cross-section plan enclosed ⁽⁴¹⁾. These really should be cored.

These slant holes ought to be drilled from the footwall side rather than the usual hanging wall side for several reasons. The dip of the Mammoth Fault contact throughout the Mammoth Mine is a quite consistent 87 degrees west, but therein a problem arises.

There were two different surveys of the 1,000 ft. level in 1925; one in August ⁽⁸⁰⁾ and a resurvey in October ⁽⁸¹⁾. Apparently a mistake of exactly ten degrees was made in a deflection at station 1008 on the August map, and a corrected map was drawn in November. This results in an ambiguity between the two maps of 83 ft. laterally in the location of the Wasp ore shoot at the 1,000 ft. level. The October resurvey is presumed correct.

If the drilling is done from the footwall side, and that presumption is wrong, no harm is done; the vein would just be intercepted sooner than expected. However, if it were drilled from the usual hanging wall side, and the presumption proved wrong, the vein would be intercepted much deeper. It might be missed altogether at the 750 ft. level. This is illustrated by the enclosed drilling plan⁽⁴¹⁾.

The footwall of the Mammoth Fault has never been identified on the Wasp claim. In the Wasp open pit, the Mammoth Fault hanging wall contact between the altered latite (dacite?) on the west and the granite on the east is sharply defined. There is a two-inch seam of fault gouge at the contact. The gold values lie eastward in the adjacent brecciated, silicified granite. The highest values lie in the first two to three feet against this contact. Then the values simply grade out eastward into the harder granite dike, without a definite cutoff.

There is a further reason for slant-hole drilling from the footwall side. In 1983, Wayne Blood, while drilling for the then operator, recovered 7-oz. and 10 oz. assays from 25-ft. depth from two vertical boreholes about 25 ft. out in the intruded granite, near the path of the proposed slant holes.

They tried to intercept these values by horizontal drilling from within the pit, but failed. These two drill holes did not define any particular structure as Blood advised there were a couple holes between them that were "down in the tenths of an ounce." However, they might be intercepted by the proposed slant holes if drilled from the footwall side. Also, the topography makes a drill setup far easier on the footwall side. There are no records of the prior operator's drilling. They had an office fire, with a loss of their records, about the time Clay Worst, the owner, called for an accounting and terminated their lease for failure to pay royalties.

There appear to be four development options for the Wasp. The first would be sinking an inclined shaft, following the ore shoot itself down the 87 degree dip. This would have the advantage of sinking on pay ore, but faces problems created by the original lessees in 1983.

The high-grade ore shoot then averaged six to eight feet in width, averaged 13 ounces of gold per ton, and was encountered under only four feet of overburden. It was five miles from town, had paved-road access, water, electricity, and an owner/lessor who was totally occupied elsewhere. It was a poor miner's dream come true. They should have started sinking a shaft on it, but chose to begin a surface mine.

Surely they knew that pursuing an eight-foot vein as a surface mine would quickly result in a stripping ratio that would end their operation, but would be immensely profitable initially. This evidently caused them to "grab the easy pickings" and not pay royalties as promised, knowing their plan would eventually be discovered and their lease terminated.

When Worst, the owner, called for an accounting, the operator reported \$200,020. in smelter returns, all from one refiner. Worst determined an actual total of \$536,369., which included sales to six other refiners, which the operators finally admitted. It may be presumed there were other sales that were never disclosed. With a shortage of \$50,452. in Worst's royalty account, he terminated their lease. None of the royalty shortage was ever paid.

Since a 20% deduction was taken for refining the "black sand" concentrates, the gross production exceeded \$670,000., taken from the top 45 feet of the Wasp pit. It may have been twice that amount.

Worst then entered an agreement with a second operator. They removed some 10,000 tons of low grade material to lower the ramp to the pit. During this process they dozed out about 1,560 tons of mixed lower grade ore. Some of this material was processed at a small mill they built on a nearby leased property.

The mill was junk. Due to heavy down time they averaged only about two tons throughput per 8-hour shift. The result was that with this primitive plant they recovered about one ounce of gold per ton, probably not over a 50% recovery, since the head ore averaged close to two ounces. They were losing the fine gold.

The tails were never assayed. The millsite was later leased to another operator who just hauled the tailings away and then abandoned his lease.

Smelter returns from the sale of concentrates and dore' during that operation were \$47,832. It was obvious that this mill could not show a profit on two-ounce mill heads. Worst also discovered that two of his then associates were involved in the non-payment of royalties during the previous operation. He terminated the second operation.

Worst was left with \$28,528 in unpaid bills, owed \$22,342 in wages, and owed for \$42,000 worth of ore shipped from the property. None of this was ever recovered. If Worst sounds a bit paranoid, it is based on bitter experience!

The Wasp was inactive from 1985 until 1992 other than sampling and assaying by Worst. The last channel sample taken across the vein at the bottom of the pit then assayed 7.68 oz. Au/ton 3.9 ft. wide, and 3.48 oz. Au/ton 8.2 ft wide.

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In May 1992, as an experiment, they set up Rex Thompson's spiral classifier, actually a placer machine, on the Wasp. No head ore was available, as the pit was then filled with water, so they ran samples of stripping waste from eight random locations around the Wasp pit. They recovered gold from every sample. There was nothing quantitative learned, since this was lode material that was never crushed or ground to liberate the gold. It only evidenced that even the stripping waste contained considerable values in the fines. A color photograph ⁽⁴³⁾ of the gold buttons recovered is included with this report.

Regrettably, all this left the Wasp shoot in the bottom of a large open pit into which the surrounding 5 acres of surface drains. Once committed to a surface operation, subsequent operations have compounded the problem by further enlarging and deepening the pit. To continue mining the original Wasp ore shoot as a surface operation is impractical, and would further complicate the problem.

Sinking a shaft in the bottom of the open pit would require some means of excluding surface water runoff. One might extend a watertight collar upward from the bottom of the pit, aligned with the 87 degree dip of the vein to a point safe from surface flooding. To extend this collar to the original surface, some 60 feet, and backfill the pit around it, would be expensive. The advantage would lie in further sinking on pay ore with its attendant cash flow.

The second development option would be to go east into the solid granite that intruded the fault and sink a vertical shaft. Then crosscut about every 100 feet to intercept the Wasp shoot, then stope. Fortunately, the vein dips only three degrees off vertical, so the crosscuts would remain short.

The disadvantage is that you would be sinking and crosscutting in barren rock, creating no cash flow. Also, you would have to leave considerable pay ore between the roof of the uppermost stope and the bottom of the present open pit, enough to support the weight if the pit were partially filled with surface flood water. A diversion dike around the periphery of the pit might help.

A third development option involves an electromagnetic survey Worst conducted with induced currents on the northeast part of the Wasp claim, near the open pit. They discovered four anomalies ⁽⁴²⁾, all within 30 feet of each other.

They had only a percussion air-track drill and 60 feet of steel. They drilled the first anomaly, and only averaged 0.10 ounces for the top 40 feet. Between 40 and 50 feet they averaged 0.70 ounces. It should be noted that a loss of 50% of precious metal values in an uncased percussion drill hole is common, due to the values getting hung up in the cavities and rough walls of the borehole. Even reverse circulation drilling with a rotary Tricone bit has resulted in a 51% loss of metal values remaining in the drill hole (*California Mining Journal*, Jan. 1988).

At 50 feet the drill hammer broke down, and they could neither continue nor withdraw the drill steel. They had the hammer repaired and managed to salvage the drill steel, but the borehole had caved beyond salvage. They do plan to redrill this anomaly, plus the remaining three. They also plan to complete the electromagnetic survey of the area surrounding the pit, as there may be other anomalies.

If drilling these anomalies discloses another ore shoot, the third development alternative might be to sink on this shoot in pay ore, and then crosscut to the original Wasp shoot at depth.

A fourth development alternative would be to set a crane at the surface on the granite footwall side, and attempt to continue mining the Wasp shoot with a clamshell.

The vein is sufficiently brecciated and friable at that point that no blasting has been required. All mining has been done with a trackhoe and skidloader. The State mine inspector advises it is permissible to clamshell to the 125-foot level. Whether this is feasible can probably only be determined by trial. Worst plans to pursue it, but with some misgivings, as deepening the surface working just compounds the mistake made by beginning it as a surface operation. The Wasp should have been an underground operation from the start.

The Wasp pit makes about 1,500 gallons of water per day. A two-inch pump handles the inflow at perhaps a 5% duty cycle. If not pumped, the static water level in the Wasp pit stabilizes at about the 20-foot level. However, it is questionable if this shallow water table existed at Goldfield prior to the construction of Mormon Flat Dam, and the resulting Canyon Lake, to the north. Nevertheless, the high gold values "in the grass roots" may have been the result of a supergene enrichment from surface waters.

However, Bedford ⁽¹²⁾, in 1923, and others, state that the 1,000 ft. level in the Mammoth Mine still had not reached what they called "the constant water table."

Following his report on the 1,000 ft. level, Bedford states, "Anticipating that the water level will be constant at about the 1,200 level, and that the nature of the mineral deposition will change as outlined under Historical Geology and that ore deposits of economical importance will be proven to be greater in extent, and better in average values than those deposits now developed, I would strongly advise that your main shaft be sunk an additional 450 feet from its present (1,000ft.) bottom."

If Bedford is correct, the true zone of secondary enrichment, with the richest ore, may lie even deeper than the 1,000 ft. level, rather than being only a grass-roots enrichment found at the surface.

A final point on the Wasp ore shoot: There were individual high grade pods encountered in the Wasp pit; these pods ran several hundred ounces of gold to the ton. They carried heavy galena and malachite (lead and copper), while the surrounding gold-bearing vein itself carried little of either. The most recent of these pods weighed about 300 pounds and assayed 244 ounces of gold and 56 ounces of silver to the ton. However, assay records of the 1,000 ft. level indicate the gold-bearing vein itself carried both lead and copper.

We have no geologist's opinion as to how these high-grade pods were created and got where they were. But if the Wasp shoot at some depth were an actual vein of this material, from which these pods were detached and relocated, and that spot could be found, it could be very profitable.

The Wasp has a complete laboratory for both fire assay and wet chemical analysis, and small batch refining. Worst is not a registered assayer, but took a course in fire assaying at the Mackey School of Mines, University of Nevada / Reno, and has college level chemistry.

There is a 3-phase electrical distribution line to the mill, but the Wasp had access to cheap diesel fuel, so we are running a GM 480-volt 60-kW generator with a freshly overhauled GMC 6-71 engine. With the entire mill running, it pulls about 30 kW, so the generator has plenty of reserve capacity for additional equipment. 120-240 volt single phase current is supplied from a public utility, but is also available from the generator. There is bulk storage for 3,000 gallons of fuel. The Wasp mill was built with new and used materials, and excellent used machinery, no junk.

The mill is designed to process about 10 tons of head ore per 8-hour shift. The mill flowsheet is a simple wet gravity process. The head ore is dumped on a 3-in. grizzly; 95% passes. The 3-inch minus is then hauled with a front-end loader up a ramp and dumped into an elevated 5.5 x 7.5 ft. bin.

We have a Syntron apron feeder for this bin, not yet installed. Ore then drops into a $6" \times 8"$ jaw crusher which takes it to about 3/4 minus, then into a 12" x 18" roller mill which takes it to about 3/8 minus.

Then up a 16" x 35' belt conveyor to a 5' x 7.5' bin with vibrator feeder. Then up a 12" x 10' belt conveyor to a 4 $\frac{1}{2}$ foot Hardinge conical ball mill with a four-foot long cylindrical screen on the discharge. We are currently using a 30-mesh screen. Oversize is passed under a magnet to remove tramp iron, then returned to the ball mill intake via a sand screw in closed circuit. The 30-minus discharge goes to a distributor box, where it is split between two Stephens (4' x 8') rougher tables, which have new sand decks. Concentrate from the rougher tables go to a single Stephens (4' x 8') finishing table, with new deck.

There are also two 3-R reverse spiral concentrators in the circuit. Rougher table tailings go to the first 3-R Spiral.

The cons from the first 3-R Spiral and the tailings from the finishing table go to the second 3-R Spiral. Cons from the second 3-R Spiral, and the middlings from the finishing table, go to a sump where they are pumped back to the feed end of the finishing table.

Tails from both 3-R Spirals go to a sluice box, then to a dewatering screw. Sand from this screw is elevated by a $16" \times 25'$ belt conveyor into a dump truck.

Tail water goes through two 1,000 gallon steel settling tanks in series, then to a 25,000 gallon mill pond with plastic liner. From there a 2-inch pump recirculates the clarified water back to the mill. Mine water supplies all needs for the mill.

A 2-inch service line from a 6-inch domestic water main at the Wasp boundary supplies any supplemental water, which is used only for cleanup.

The roller mill and ball mill are in new condition; the jaw crusher, conveyors, tables, and other equipment are in excellent shape. All motors except the tables and 3-R Spirals are 480 volt, 3-phase, powered by the generating plant. The three table motors are 240 volt, single phase, and can be operated on either public utility power or the generator. The 3-R spirals have DC motors with rectifiers and speed controls, which operate off one leg of the 3-phase generator. Each sand screw has its own independent hydraulic motor, pump and controls. Almost all wiring is in conduit, most of it underground.

Concentrates from the finishing table are amalgamated, the amalgam retorted, and a dore' bar poured. The Wasp refinery can produce a 995 fine gold bar, but chooses to market the 750 fine dore'. The smelter charges only 4% for refining the Wasp dore'. Since the Wasp cannot hallmark, even their 995 fine bar would entail this 4% refining charge, so refining to 995 fine would be pointless.

The Wasp mill's capacity is limited by the ball mill, which the manufacturer rates at one ton per hour, grinding ¹/₂-inch hard quartz to 100 mesh. However, Wasp ore is very easy milling, the ball mill feed averages finer than ¹/₂-inch, and already has a high percentage of fines. The mill is currently fed 1.2 tons per hour, which seems a bit under capacity.

The mill is just being placed in production as of this writing, so no current production figures are yet available.

The Wasp owners have never solicited nor accepted investment money from anyone, which eliminates any possibility of adverse claimants against their clear title.

Regarding an exploration license and purchase option, the owners would require the three drill holes mentioned earlier, aligned to intercept the Wasp ore shoot at it's known point of highest assay. Calculations for this alignment will be agreed upon prior to any license. After these three boreholes, the developer may drill anywhere he wishes.

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The Wasp was bought in 1978 by Worst and an associate, Cliff Sovig. They divided the property. Worst took the north half, upon which all existing development was later done. Sovig took the south half. Sovig died, and the south half is owned by his successors. The south half is undeveloped land, mostly dacite, however the Mammoth Fault as platted ⁽⁷⁷⁾ does cut across a portion of it. The owners believe the property is worth more if sold as an integral unit, and it is presumed that a developer would want the entire claim.

Residential land in the area, with utilities available, runs from \$30,000 to \$50,000 per acre, some as high as \$100,000 per acre. This can be verified by contacting any area real estate broker.

The owner's residence was FHA appraised five years ago at \$92,000. and is currently rated for insurance purposes at \$140,000. With the mill, laboratory and outbuildings included, just the surface estate of the claim ought to be worth \$750,000.

If drilling indicates there actually is \$11 million in the Wasp ore shoot, the entire property ought to be worth the asking price of \$2.5 million. This is attributing only \$1.75 million to the mineral estate, as the buyer would still have the remaining value of the surface estate after the mine is depleted, which value should by then be substantially appreciated.

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Some speculative value is attributed to the fact that in 1893-1895, within pistol-shot distance, \$34 million (at today's gold prices) was taken out of two ore shoots on this same geological structure.

If a drilling program indicates the property is worth less than the sellers now presume, they may consider a downward adjustment in the price. Such adjustment would be solely the sellers' prerogative. The sellers will not index the price to drilling results in advance, as this excessively transfers control of the price to the buyer. If the developer abandons the project after the drilling program, he retains no residual earned equity in the property.

The owners are not soliciting lease offers, having already sustained losses of \$143,000 from the infidelity of lessees. In any lease event, the existing Wasp ore shoot may not be additionally mined by surface methods without the owner's consent, the granting of which is questionable. Neither will the owners accept any investment money or sell any fractional interests.

If a buyer/developer has other ideas, the owners will listen, but may not depart far from the outline offered here. Meanwhile, until someone makes a proposal, the owners will continue with exploration and development, during which the price and terms may be adjusted either way depending on the results of the work.

The owners would consider reducing the \$2.5 million "asking price" by \$140,000., and retain only the surface estate to the 1¼-acres containing the owner's residence, with an easement for ingress and egress. This 1¼-acres is across the highway from, and would not interfere with, the mining operation. The mineral estate underlying this 1¼-acres would be conveyed to the buyer, who must warrant against damage from blasting or subsidence. This offering, however, is not conditioned upon such an arrangement. If the buyer requires the entire property, the sellers will need a 6-month leaseback on the owner's residence to permit their orderly departure from the premises.

Clay Worst

1,200,000 440.000

STANDARD OPERATING PROCEDURES

THE OLD WASP MILL

Note: All components of the mill must be properly inspected and serviced prior to startup.

STARTUP:

1) All switches OFF on all electrical panels *EXCEPT* SRP 120-240V disconnect on Panel No. 1, and 240V 3-phase disconnect on Panel No. 3, which serves mainly to fuse Panel No. 4 and as an emergency disconnect.

2) Start generator; set at 60 cycles/sec., 450V; 100A circuit breaker ON.

3) Panel No. 1: Verify that all personnel are clear; Main Disconnect ON.

4) Underground city water value at east end of decantation tanks OFF; this value is *normally* never opened. It supplies the yellow standpipe by electrical panel No. 3, which is ordinarily never used.

5) 2-in. valve on yellow standpipe at east end of decantation tanks ON.

6) Tailings conveyor ON.

7) Tailings sand screw ON.

8) Hydraulic oil radiator cooling fan ON.

9) No. 1 (mill supply) pump ON.

10) Finishing table main water valve ON, motor ON; during remainder of startup, finishing table con line will vary widely and must be *constantly* monitored.

11) Finisher spiral motor ON.

12) Middlings sump pump ON.

13) Rougher spiral motor ON.

14) No. 1 (east) rougher table motor ON; water supply valve ON; brush any residual slimes from deck.

15) No. 2 (west) rougher table motor ON; water supply valve ON; brush any residual slimes from deck.

16) Main water valve to concentrating circuit fully ON; this may have been left partially closed after previous shift's cleanup.

17) At ball mill "Christmas tree": Hose valves color coded blue and yellow OFF; valve to ball mill intake scoop valve (color coded red) fully ON. Fine adjustment at intake scoop is made *only* at valve at intake scoop spray bar. Normal setting is with red marker on valve handle cracked open to between 2:30 and 3:00 o'clock position.

18) Main water valve to ball mill (color coded orange) ON.

19) Ball mill ON.

20) Ball mill feed conveyor ON.

21) Ball mill feed conveyor vibrator ON.

4

22) While waiting for ball mill to load, wet-brush table tops.

23) When ball mill begins to discharge, turn classifier spray bar 1 $\frac{1}{2}$ inch valve (color code purple) ON.

24) As ball mill loads, monitor oversize discharge launder; at Christmas tree valve (color code yellow), add water *intermittently only as needed* to prevent plugging launder discharge until mill stabilizes. Then only *minimum water* to maintain transport of oversize to sand screw. Sand screw drain fitting leaks until slimed up; *do not repair*, as this initial water leak is *beneficial*.

25) When oversize in sand screw accumulates nearly to level of intake pipe, turn sand screw ON. If excess water has accumulated during startup, and water will not transport, *lock sand screw out*, open hopper guard and bail excess water into hopper discharge launder. Close and secure guard. Start sand screw.

26) If mill discharge slimes sufficiently to overload classifier screen, turn feed belt OFF until slimes diminish, usually 2 to 3 minutes; then feed belt ON. This is common occurrence with 30-mesh grind; only occasionally with 20-mesh grind.

27) If oversize discharge becomes too thick (sandy), increase water (valve color coded red) to ball mill feed scoop. Excessively thick oversize will cause mill to discharge tramp iron.

28) Periodically check permanent magnet in oversize discharge chute for captive tramp iron. Remove plastic cup and magnet *together*; then remove magnet from cup to allow captive metal to release; replace magnet in plastic cup and replace in discharge chute. Do this quickly, as any tramp iron discharged while the magnet is removed will pass into and *may plug the sand screw*. This *can* be completed in 3 seconds.

29) At Christmas tree: Water supply valve (color code blue) to nozzle in oversize sand screw *discharge* chute back into the ball mill is *almost never required*. Use *only* in the extremely rare event the chute becomes plugged.

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SHUTDOWN:

Note: Begin shutdown sequence one hour before shift ends.

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1) Have wash hose connected to city water, spray nozzle attached, and located near ball mill.

2) Vibrator feeder OFF.

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3) Ball mill feed conveyor OFF. Finishing table con line will vary widely and *must be constantly monitored* during remainder of shutdown sequence.

4) Allow mill to digest. After approx. 10 minutes, turn OFF ball mill feed scoop water valve (color code red). During this time, use minimum water to oversize launder.

5) Continue to digest until water ceases to discharge from mill (approx. another 5 minutes). Then, at Christmas tree: $1\frac{1}{2}$ inch valve (color code purple) to classifier spray bar OFF, $1\frac{4}{2}$ inch main valve (color code orange) to grinding circuit OFF.

6) Ball mill *starter switch* OFF; if necessary, tweak starter switch to stop mill with any of the four open screen sections of classifier cage at bottom position (facilitates spray cleaning of classifier screen). *Then* ball mill disconnect OFF. *LOCK OUT BALL MILL* <u>DISCONNECT</u>. Never rely upon locking out ball mill at the starter switch, because a short circuit in the starter could accidentally energize the mill.

7) Reduce concentrating circuit water pressure to 5 psi; valve is located next to water pressure gauge.

8) Spray-wash classifier screen, launder and discharge pipe.

9) Spray-wash oversize launder and discharge pipe.

10) With sand screw still running, remove sand screw drain plug, back-flush through drain fitting, and drain contents. Place hand over drain fitting to slow the discharge of sand, or drain pipe will plug. Use sufficient flush water to avoid plugging drain pipe.

11) Oversize sand screw OFF. Remove lock from ball mill disconnect and <u>LOCK OUT</u> <u>SAND SCREW DISCONNECT</u>.

12) Open guard on sand screw hopper. Spray-wash entire sand screw flight back into hopper and drain. Remove any remaining tramp iron & rocks from sand screw and from screen in drain sump. Close and secure guard. Replace drain plug.

13) Spray-wash distributor box and both rougher table supply pipes.

14) Spray-wash first rougher table feed box/and deck. *Be gentle*, especially on upper riffles; *don't blow the gold off the table*.

15) First rougher table motor OFF; water supply valve OFF.

16) *Gently* spray-wash and brush first rougher table deck; spray-wash tailings launder and pipe; *gently* spray-wash con launder and pipe.

17) Repeat steps 14 through 16 on second rougher table.

18) Readjust water pressure to 5 psi (it will have increased as rougher tables were shut off).

19) Rougher spiral OFF.

1. 6 ×

20) Middlings sump pump OFF.

21) Spray-wash finishing table feed box.

22) Gently spray-wash first few riffles of finishing table; observe gold moving down.

23) Gently spray-wash lower riffles of finishing table; spray-wash tailings launder & pipe.

24) Middlings sump pump ON. Open middlings return valve at finishing table feed box one additional turn. Stir and wash contents of sump into sump pump.

25) When gold line disappears on finishing table, turn finishing spiral OFF; continue cleaning sump until concentrate line essentially disappears.

26) Middlings sump pump OFF.

27) Spray-wash finishing table feed box.

28) Gently spray-wash finishing table deck.

29) When no more gold is being recovered, turn table motor OFF; water supply valve OFF.

30) Gently spray-wash and brush finishing table deck.

31) No. 1 (mill supply) pump OFF.

32) Open tailings sand screw guard; secure with bungee cord.

33) Disconnect and remove sluice box extension and diverter.

34) Use submersible pump to dewater sand screw hopper into No. 1 decantation tank.

35) Carefully hand any feed remaining sand into sand screw auger.

36. When sand screw hopper is empty, turn sand screw OFF.

37) Hydraulic oil cooler fan OFF.

38) Tailings conveyor OFF.

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39) Panel No. 1 main disconnect OFF.

40) Generator OFF; close side doors and close exhaust stack cover.

41) Secure generator enclosure.

CRUSHING CIRCUIT ONLY:

1) No. 2 (crushing station) panel disconnect ON.

2) Belt conveyor ON.

3) Roll crusher ON.

4) Jaw crusher ON.

5) When jaw crusher is up to full speed, begin crushing.

6) While crushing, frequently monitor output of conveyor belt.

SHUTDOWN:

Notes as

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1) When jaw crusher is empty, switch crusher OFF.

2) When roll crusher is empty, switch roll crusher OFF.

3) When belt conveyor is empty, switch belt conveyor OFF.

4) Crusher disconnect OFF.

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Wep of Miner Muneral Resources 6-23-90 Phoeney anyona Wear Serv I received your letter stateing that you have no reference to a gold minering effort by Western Fabs and enquicesing some times ago it was known as mom interprises and you thave it under that mence I am sending his business card and mine advess and also a coppy of an article he sent me published by the Sugering all of this records. This in my opinion will probally put him out of business which I think some one has ben trying to make this line make this happen for some time because of the problems that he has had trying to get started milieing I know Paladon people shave brought suits agenst him but he won it is the time loss and costs that has hurt and I also seen where the state wanted to make a parfor inthe a & bought gravel from him to be proceeded for the gold Content and I bought crushed stone to build a drive way both purchased for a benefit to me and purchased with a bill of sale I don't see how that is selling fraudelant and after two visits of one weak sof each he says there is gold securities I pick and shovel miners got wealthy mencing this property when gold was 15 of modern equipment and improved methods should do well miliening 350 of this is how dree it thank you for the letter JUN 2.6 1990 Amold & King 1208 mich. ave MINERAL RESOURCES alma mich 48801

5-14-90 May 17 10 16 AH '90 90 - 29 157 Dor. Rose mofford Dear Madam ' I am writing in regards to the gold mineing effort Hestern Laber Engineering as I am an investor in chis project I have geoligist reports that there is gold on this Property and they are quite favorable with other gold mineing ventures that I have invested in that are mineing gold profitabley I moke sense to me that if this mine produced in the quanties when gold was 20 dollars prog that with modern methods and gold over 350 prog that it still can be a profitible venture I am of the openion that money is being spent to Put obisticles in his way to present him from making a success in this project With investors in mind theaply way they Can hope to recover their investment is for this mineing operation to Continua Un something ic going on that I am not aware of why don't the state defer his taxes for one year and and give him a chance to get on with his operation I know this has been done before

thank you

louold I. Keng 1208 mich aver

alma MT 40001-1221

GOLDFIELD MINES, INC. MINES, INC. Box 37 Apache Junction, Arizona

May 2nd, 1951

Department of Mineral Resources Mineral Building, Fairgrounds Phoenix, Arizona

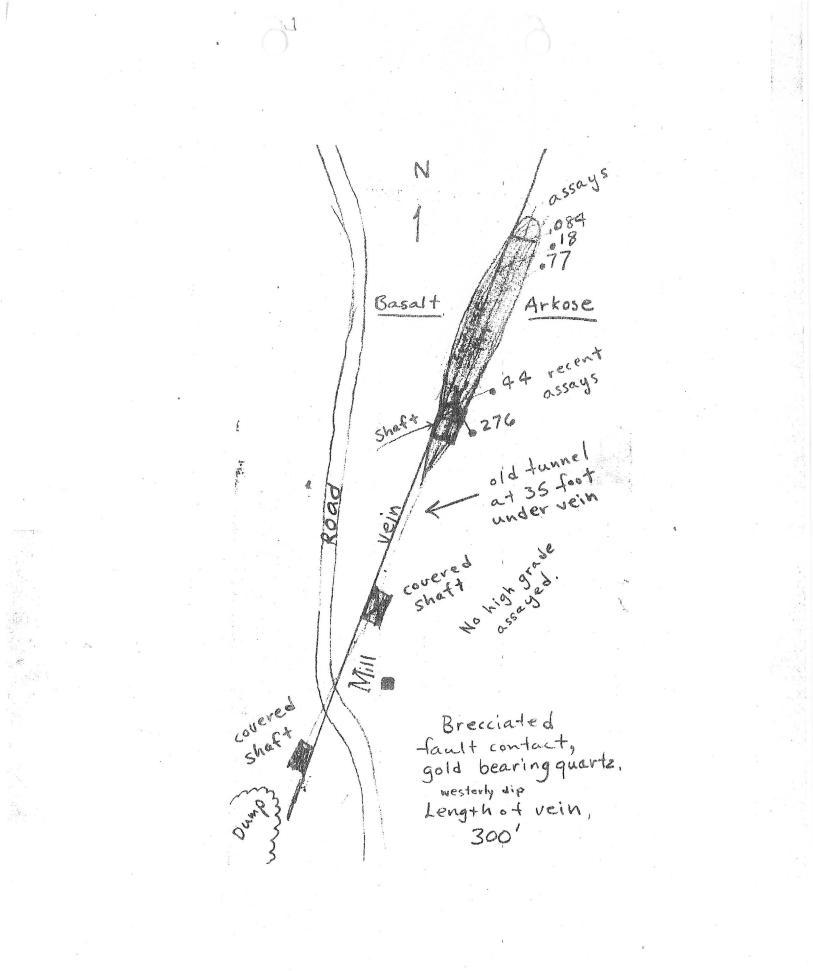
Gentlemen:-

Please send us four copies of Form MF-100, for application for Serial Number for Goldfield Mines, Inc.

If you cannot furnish us with four copies, please send us one copy and we can make the extra copies needed.

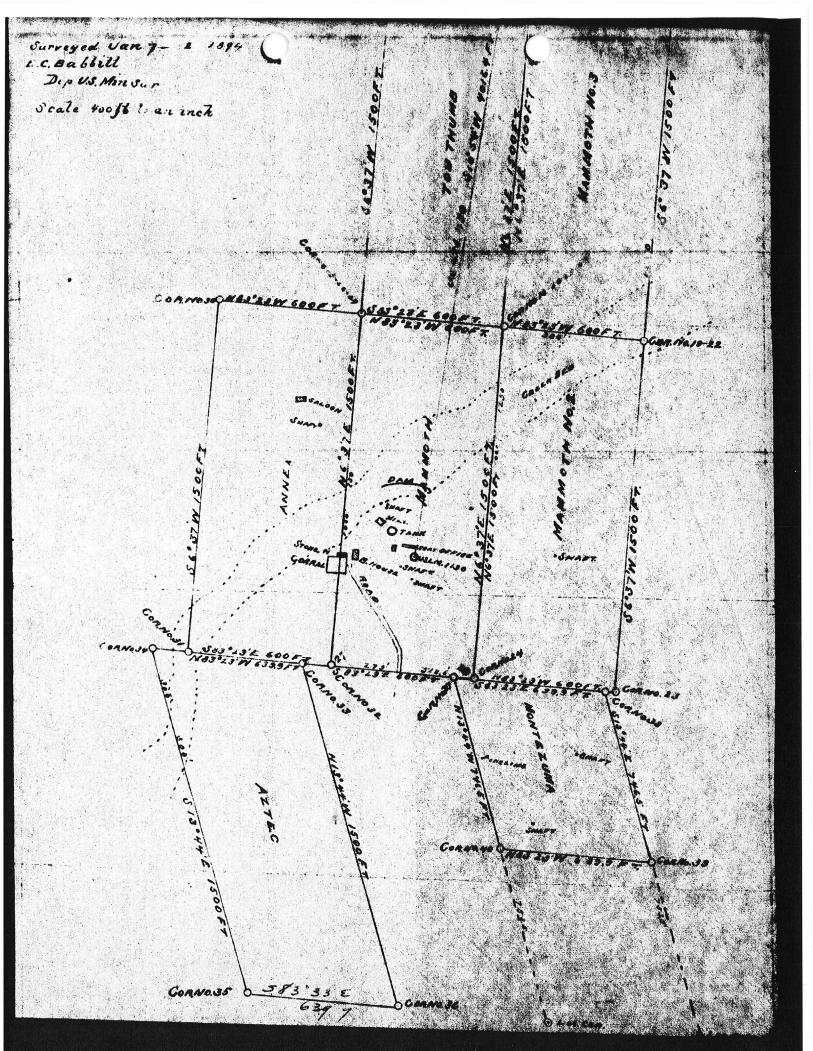
Very truly yours,

✓ GOLDFIELD MINES, INC. V Jon J. Quesell Secriy-Mgr. Q.a.



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BLACK QUEEN LODE MINING CLAIM SUPERSTITION MINING DISTRICT By Geologist John D. Wilburn

HISTORY

Early production records of the SUPERSTITION MINING DISTRICT have eluded researchers, but it is common knowledge that the district's largest mine, the Mammoth, produced at least one million dollars from the "Mormon Stope" that was mined to a depth of 265 feet from the years 1893 to 1897. The Black Queen Mine is the districts second largest mine found only 3,000 feet to the north of the Mammoth with depths of only 150 feet known, and drifts along the vein. Early production on the Black Queen is known to have been approximately a quarter million dollars. Later sporatic activity on the Mammoth Mine yielded a known production of \$67,000 between 1913 and 1925. Some mining on the Black Queen Mine in the late 1920's and in 1930 was carried on, production is unknown. It is known that the mine was closed in this period due to high-grading and the property has remained idle since.

GEOLOGY

The vein in the Black Queen Mine occurs along a fault contact between basalt on the west and arkose on the east. The quartz veins are found entirely in the arkose and is traceable along the surface for some 300 feet, all of which carries free gold, which on the whole is very fine. Several faults along the contact dip steeply to the west and have produced a breccia zone in the arkose. Adjoining the basalt is a band of kaolin a foot or so in width, next a one foot bank of red crushed quartz, a product of oxidized pyrites and post faulting, and several feet of quartz in brecciated The width of the vein varies along the strike, slightly arkose. east of north to south, from a foot to six feet. This breccia zone is traversed by many quartz veins from narrow seams to those more than a foot in thickness. Free gold forms relatively large patches of yellow within the quartz. The richest ore occurs where there is an abundance of black manganese with the quartz usually with some rhodonite. Spotty, red oxidized pyrites are found sparingly scattered throughout the quartz. The interstitial brecciated arkose in the immediate vicinity of the ore is stained with manganese, and silicified where upon the fresh red color becomes white or slightly greenish. No sulphides occur with the ore, all is free milling. Ore deposition falls into the epithermal class of veins constituting one of the richest of ore deposits. The term "Bonanza" was derived from epithermal veins such as the Comstock Lode, Cripple Creek and Goldfield, Nevada.

PRESENT PERSPECTIVE

The Black Queen Mine is the districts most promising property. The fact that the ore is very rich and free milling and the fact that the mine has not been mined to any great extent offers excellent prospects of developing a valuable gold mine. (Presently, rich ore has been found only inches below the surface several feet wide and 100 feet long. A fifty ton mill and concentrating tables are upon the property at this time, ready for milling.) Ore was found to depths of 1,050 feet in the Mammoth Mine. Faulting and breccia in the arkose on the Black Queen is the same as in the Mammoth and such strong faults should carry auiferous quartz to depths of at least 1,000 feet here also. Below the 150 foot level there exists about 800 feet of virgin gound. It is reasonable to assume that within this virgin ground, several rich ore shoots will be encountered. Typical of epithermal veins, exceedingly rich ore shoots frequently occur especially with breccia structure. Prospects for another "Mormon Stope" are likely on the Black Queen.

A parallel fault to the Black Queen Vein has produced a sheeted zone to the east constituting 140 feet of narrow quartz veins. This is all virgin ground and the drill proves gold at various depths. This zone is not considered to be of economic importance from the surface, although further exploration may prove ore.

- 2 -

RECOMMENDATIONS

by

Charles B. Broan, Superintendent

ALL THIS EXTRA WORK can be done with the SAME amount of OVERHEAD expense we are under at the present time. By putting the CAGE in the South Compartment of the MAIN working shaft we can do ALL of the developing from the 800 UP and at the SAME time by using a BUCKET in the center Compartment we can sink the shaft, and if at any time we wish to stop sinking, we can, by installing another cage in the center Compartment handle a LARGE tonnage of ORE.

With the present hoist and equipment we have, there would not be speed enough to handle the tonnage. Therefore at this time I would recommend ELECTRIFYING THE MINE, for with the Present shaft -- which I can safely say is as good as any in the entire country, by electrifying everything we can use any speed safety would allow.

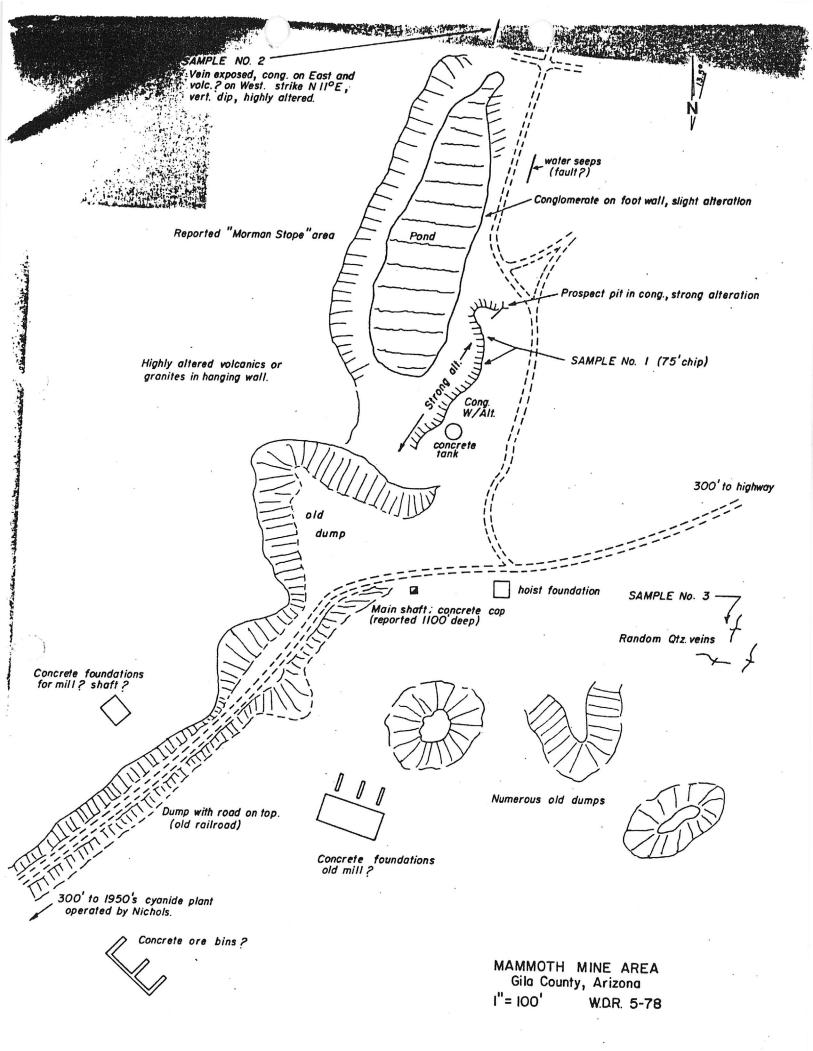
I have examined the Old Mill thoroughly and conclude that a NEW mill should be installed -- one that is up to date in every way because when we start milling we do not want any delays in the way of "breakdowns".

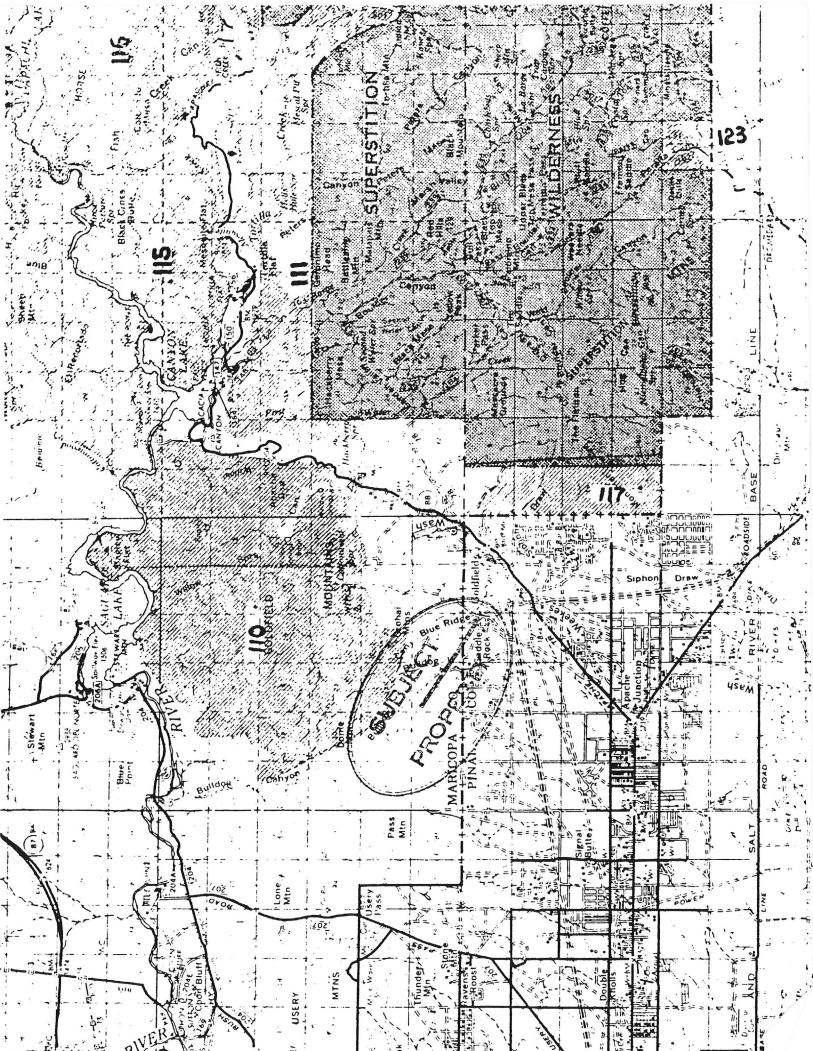
Water is the essential question just at this time, but I believe by piping from different water holes and sources of supply we have at the present time since the rains, we can find quite a supply which will be of material assistance until such time as the Main shaft can be sunk this additional footage, at which time, I believe all the water necessary will be secured, for the reason that as yet we have NOT REACHED THE WATER LEVEL, AND THIS MUST BE ATTAINED TO ENABLE ONE TO KNOW MORE DETAILS BOTH AS TO WATER AND THE PRIMARY ORE ON THE MAMMOTH CLAIM.

I further recommend the development of the Black Queen property JUST AS SOON AS POSSIBLE, for as I have said -- it UNDOUBTEDLY IS A TREASURE HOUSE. It would be advisable to repair the shaft and get in readiness for operation at once, as all assays show good paying ore, of which there is a BIG TONNAGE of HIGH GRADE and which is practically all in VIRGIN GROUND.

Respectfully submitted,

Charles B. Broan Superintendent





IRON KING ASSAY OFFICE ASSAY CERTIFICATE

BOX 14 - PHONE 632-7410 HUMBOLDT, ARIZONA 86329

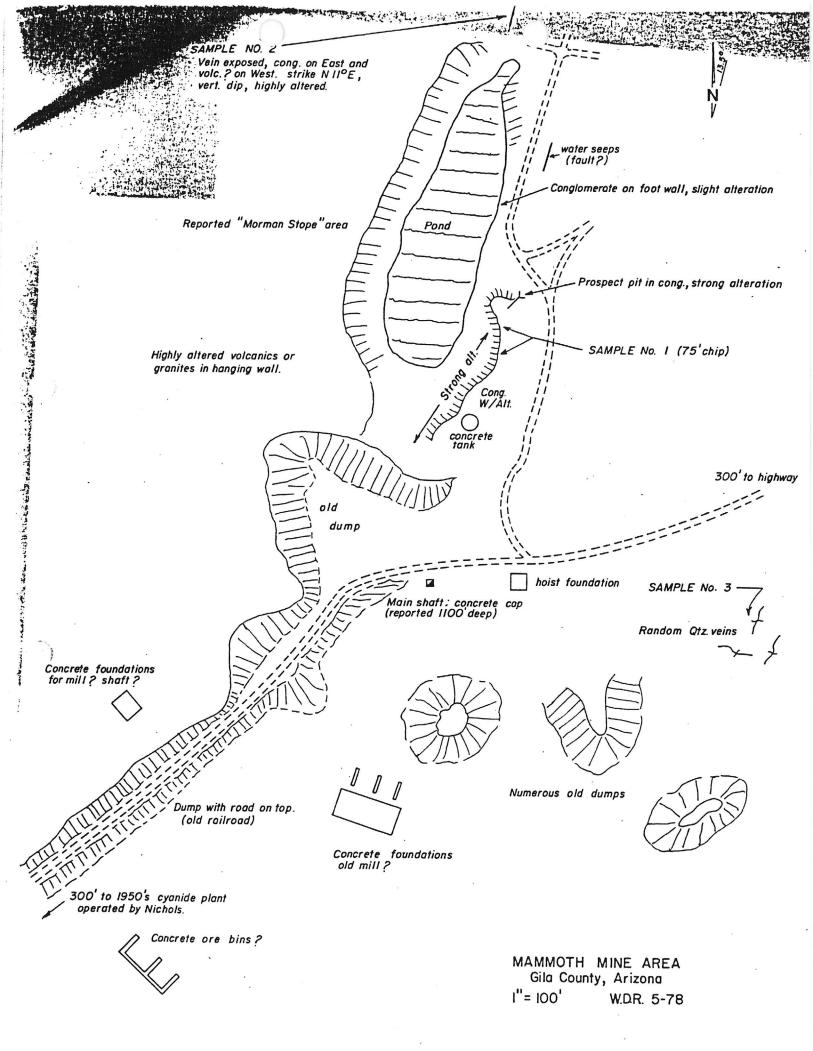
CLARK OLIVER MINING CO. INC.	HUMBOLDT, ARIZONA 86329						
SSAY Frank Clark							
ADE 6942 W. Olive, Sp. 68							
or Peoria, Ariz. 85345							
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- 48 Mesh	42.40	160.04	38.50	110.46	39.48	29.98	
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PICTURES

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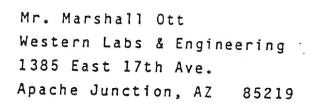
MARVEY W. SMITH, E.M. PRESIDENT

Registered Mining Engin U.S. Approved Title Abstructer

U.S. Mineral _vor **Registered Land Surveyor**

4310 North Brown Avenue / Suite 3 Scottsdale, Arizona 85251 Tel. 602 / 946-3996

November 27, 1989



Dear Marshall:

Enclosed is a copy of the plat we have drafted showing locations of samples that we cut on the Mammoth and Mammoth No. 2 lodes. Twenty-two additional samples were taken from underground on the Black Queen lode. The widths of the samples are quite variable, generally 2 to 3 feet wide, but as narrow as one inch on some quartz veins.

If this plat looks satisfactory let me know, I will stamp it, and send you additional copies. If you have any questions, please call.

Thank you for asking us to assist on this project. statement is enclosed.

Sincerely,

Harvey W. Smith, E.M.

President

HM/hm Enclosure



DEL TIERRA ENGINEERING & MINING CORP.

HARVEY W TH, E.M. PRESIDENT

Registered Mining Engineer U.S. Approved Title Abstracter

U.S. Mineral Surveyor Registered Land Surveyor

4310 North Brown Avenue / Suite 3 Scottsdale, Arizona 85251 Tel. 602 / 946-3996

November 27, 1989



Pd. - C1700175/

Western Labs & Engineering 1385 East 17th Avenue Apache Junction, AZ 85219

For professional engineering services pertaining to sampling of the Mammoth lode:

1,013 miles @ \$0.60/mile 96 hr. @ \$35/hr. (Sampling)

> Total Due Received on Account

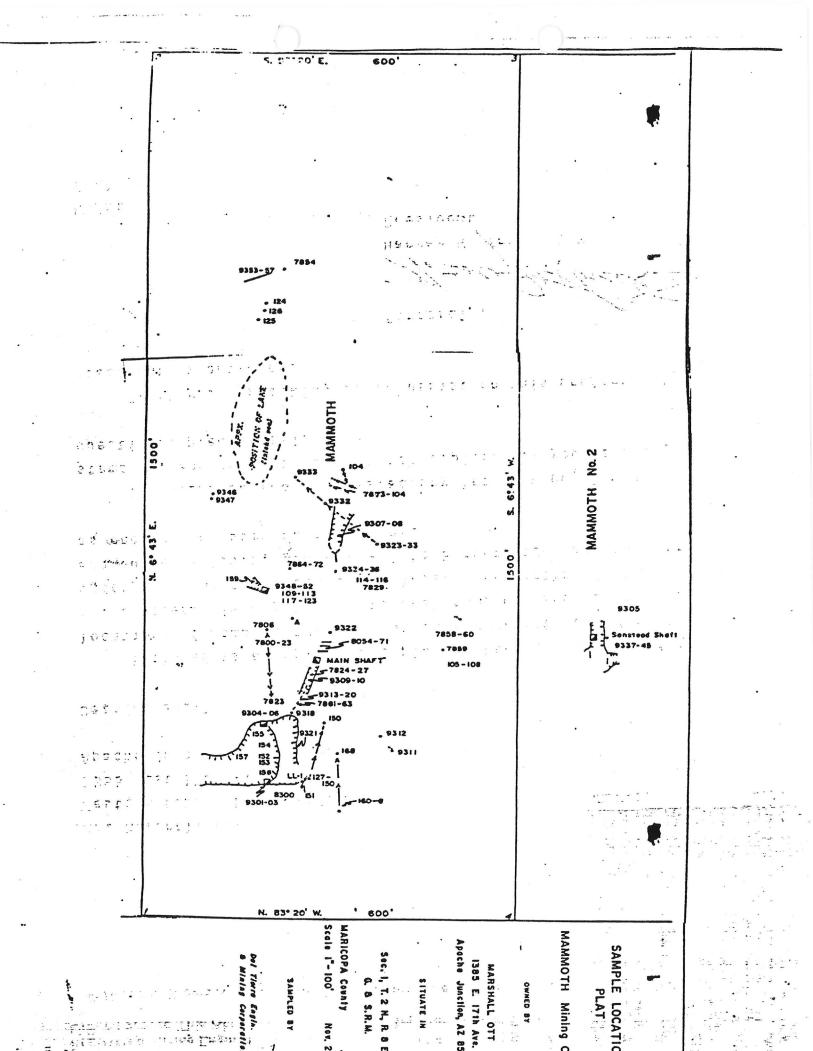
> > Balance Due

607.80 \$ 3,360.00

\$ 3,967.80 2,750.00

Respectfully submitted

Harvey Śmith, E.M. W President



15 S. 10th AVE.

Jacobs Assay Offi Registered Assayers

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PHONE 622-08

Certificate No Z	81			TUCSON, ARIZON	A 85713	1/21/89	
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Certificate No

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Jacobs Assay Offis Registered Assayers

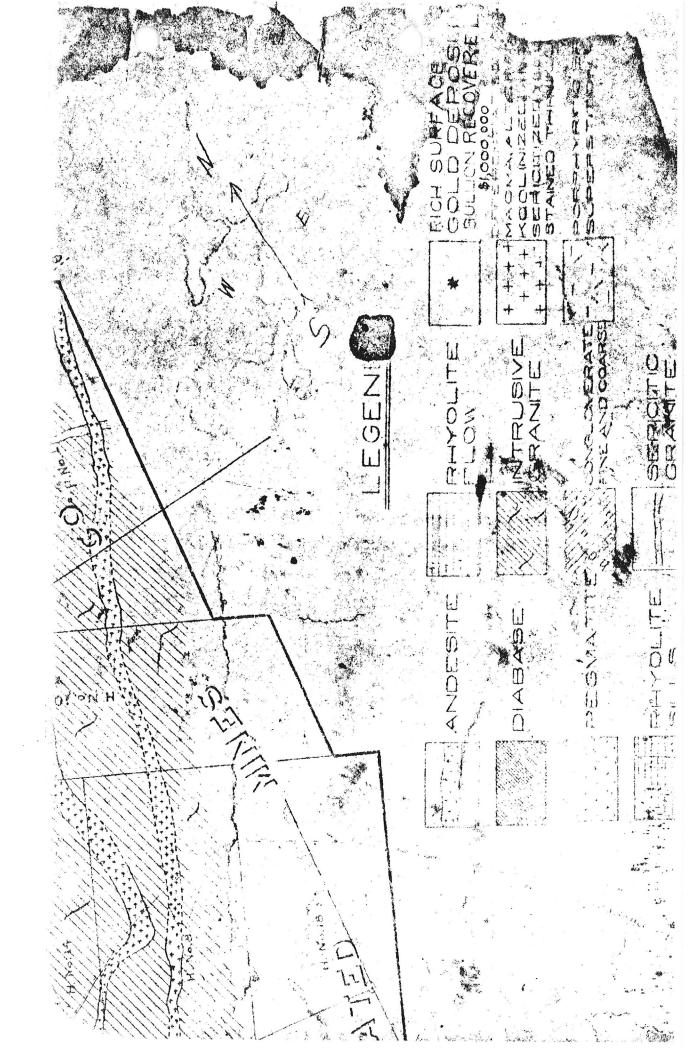


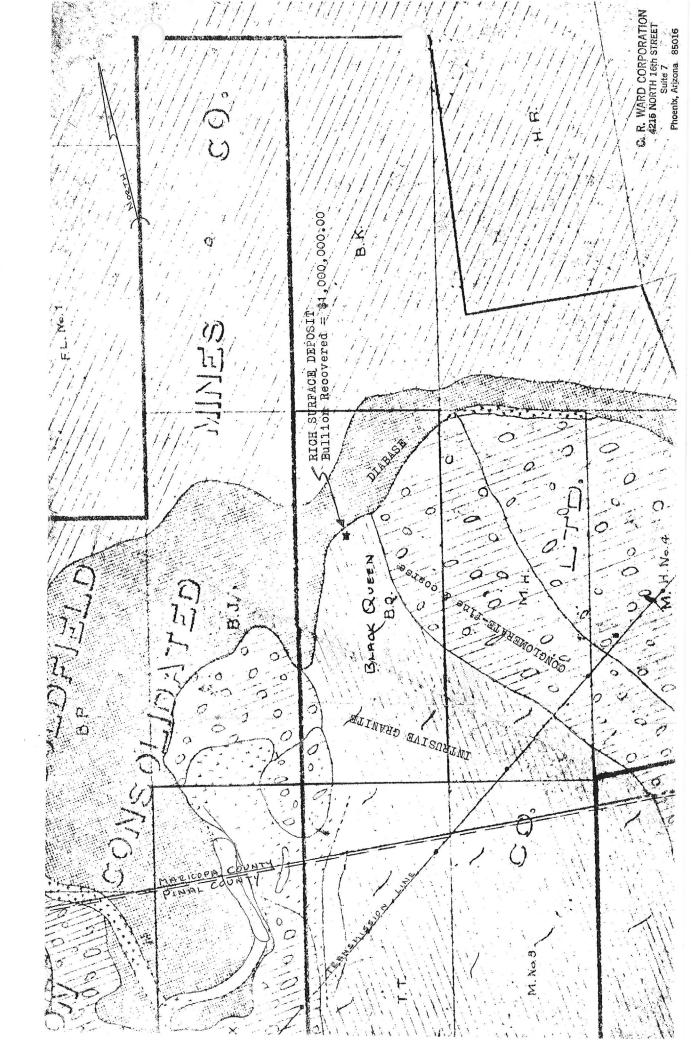
PHONE 622-08

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TUCSON, ARIZONA 85713 11/22/89 19

SAMPLE	GOLD COLD	SAMPLE MARKED	GOLD OZ.PER TON
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Registered Assayers



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Certificate No

Sample Submitted By Mr. MARSHALL OTT

TUCSON, ARIZONA 85713

11/24/89

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Certifizate No

TUCSON, ARIZONA 85713.....11/28/89......19 ...

Sample Submitted By Mr.______MARSHALL OTT

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AR-033060

UNITED STATES DEPARTMENT OF THE INTERIOR BUREAU OF LAND MANAGEMENT Phoenix District State______Arizona

Foragan Lands Linerals Records

MINERAL REPORT

Mineral Patent Application

of

Goldfield Mines, Inc.

for the

Mammoth Lode Mining Claim

(Title)

LANDS INVOLVED

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1964 AUG 31

Pinal County, Arizona Gila and Salt River Meridian T. 1 N., R. 8 E. Sec. 1: Part of StNWt & NtSWt containing 20.609 acres

LAND OFFIC DENIX. ARIZ

August 27, 1964 (Date)

· By

Lewis S. Zentner Valuation Engineer (Mining)

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Orig: Case file cc: T&R cc: PDO-M cc: USGS

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Introduction:

The Mammoth lode mining claim, located in the Goldfield Mining District, is approximately four miles northeast of Apache Junction. Arizona State Highway 88 traverses the southeast part of the claim, and other parts of the claim are accessible over dirt roads.

Field identification of the claim was made from the mineral survey monuments located at the corners of the claim, and from the section corner common to sections 1, 2, 11 and 12, T. 1 N., R. 8 E., G&SRM.

During the course of the examination carried out intermittently during the months of May and June, 1964, interviews were held with the following people:

> Mr. Hugh Nichols, President, Goldfield Mines, Inc. Mrs. Beatrice W. Lewis, resident on adjoining claim Mr. Harvey Smith, U.S. Mineral Surveyor

Mr. Harvey Smith accompanied the mineral examiners Lewis S. Zentner and Robert T. Mitcham, July 20, 1964, on a joint examination of the claim.

Topography, Vegetation and Climate:

The land within and surrounding the Mammoth claim is rolling and slopes gently to the southwest, the elevation being about 1500 feet above sea level. Overburden consisting of rocky soil, varies from a few inches to several feet thick. The area is entrenched by numerous ravines having near vertical banks. Where the soil mantle is heavy, these banks may rise from 10 to 25 feet above the washes.

The water level in the vicinity is not known, but there is water standing in the old workings on the Mammoth at about 20 feet below the surface. On the Bulldog mining claim, approximately one mile west of the Mammoth, there is water standing in a shaft 70 feet below the surface.

Vegetation is scanty, consisting of cacti, palo verde and mesquite.

The climate is mild during the winter months and hot in the summer. Annual precipitation averages 9 to 10 inches.

General Geology:

Almost all of the area is covered by a soil mantle of variable thickness. Where the mantle has been removed by stream action or mine workings, granite and volcanic rock have been exposed.

p.5.

To the north and west of the claim, Precambrian granite overlain • by north dipping Tertiary tuff and lava flows crop out.

Local Geology and Mineralization:

Granite is exposed in the cut shown on the mineral survey plat as improvement No. 2 and in the large pit to the north of the main shaft (No. 7 on plat). The granite in the cut is highly decomposed and stained red in the upper part of the cut. The red coloring is probably due to exidation of pyrite associated with the granite.

A fault zone trending N. 5° E. and dipping vertically is exposed in both the north and south end of the cut; this zone was examined and sampled. Quartz stringers were observed associated with the granite in the fault zone. The rock along the zone is highly fractured.

According to Arizona Bureau of Mines Bulletin 137, indurated conglomerate and granite breccia are also found in the area and the gold present is associated with iron oxide. Gold mineralization is found associated with the quartz stringers along the fault. In all probability gold bearing solution moved upward along the fault and deposited the gold. The vertical extent of deposition is not known but gold was reported in the deepest drift 1000 feet below the surface.

Sampling:

Samples were cut at the north and south end of improvement No. 2 (see mineral survey plat for location). The samples were assayed for gold and silver. Table No. 1 shows the pertinent data for these samples.

Assay No.	:Sample widt	n:Value Cold : per ton	:Value Silve :- per ton	r: Location
H-1	3.5 ft.	\$91.70	\$1.60	North end of Improvement No. 2
M-2	1.0 ft.	14.70	2.00	Protuall side south end of Improvement No. 2
M-3	1.25 ft.	16.10	1_80	Hanging wall side south and of Improvement No. 1

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The face at the south end of the cut is 3.25 test uide. The central part (1 foot wide) appeared massive and barren and was not sampled.

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The assay value for the sample taken at the north end appears high and not representative of the fault zone.

A second sample was cut here at a later date and panned. A few minute particles of free gold were recovered from the sample. This free gold was probably deposited as erosion of the vein occurred. The light material was carried away and the gold deposited. These high values could not be expected to persist at depth.

The assay values of samples Nos. M-2 and M-3 are probably more representative and could be expected to continue at depth.

Mining:

Mining from the Marmoth lode mining claim has been carried on intermittently since 1898. The most recent mining was done in 1950 and 1951. Mining during this latter period was by open pit. During this period of production (1950 and 1951), the company received in excess of \$19,725 in payment from smalters for gold ore shipmants. The total tonnage of ore represented by these payments is not known.

Early production records for the mine are not available. However Arizona Bureau of Mines Bulletin 137 records that from 1910 to 1930 during intermittent operations \$67,000 worth of gold and silver were recovered from 7100 tons of ore.

Mining Improvements:

Records indicate three shafts on the claim (one sunk to a depth of 1000 feet) and thousands of feet of drifts. Because of water and most of the workings being caved, the underground workings are inaccessible. It is obvious from surface workings that in excess of \$500 has been spent in developing the claim.

References:

Arizona Bureau of Mines Bulletin 137.

Lewis S. Zentner Valuation Engineer (Mining)

Attachments:

Assay Report, Ariz. Testing Lab Lewis report on Mammoth Mine Plat

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Silver figured at S 1.00 per ounce GOLD SILVER PERCENTAGES LAS. NO. ICENTIFICATION SZ. PERTON VALUE SZ. PERTON VALUE 157783 ₽M-1 2.62 \$91.70 1.60 \$1.60 157784 €M-2 0.42 14.70 2.00 2.00 157785 ₽M-3 0.46 16.10 1.80 1.80 RECEIVED Jil: 23 1954 12014 0 TERED 103.9 CIST. MGP -0-.... CLAUDE E. . . MCLEAN se. REMARKS JY:Y' Respectfully submitted. ARIZONA TESTING LABORATORIES



ETT MINERAL MANAGEMENT



n. F. HEWLETT Phone (619) 242-8297 242-5757 242-5450

March 15, 1985

15110 Mandan Apple Valley California 92307

Bill Morton P.M. Recovery, INC.

Dear Mr. Morton:

Assays collected by myself and Shane Lohse on the "Apache Junction Tailings" project were analyzed by MONITOR GEOCHEMICAL LABORATORIES with the following results:

"Don's Tailings"	Site	A		.319	troy	ounces	gold/ton	.35	A
"AP JCT BUCKET C	Site	B	;	.239	troy	ounces	gold/ton	.50	A
"Bucket C/Loc. C	Site	С.	o*	.215	troy	ounces	gold/ton	.10	A
Loc. I-12	Site	D.		.176	troy	ounces	gold/ton		
AP JUNCT BUCKET	Site	Ε		.168	troy	ounces	gold/ton	.20	٨
Loc. T-12	Site	F	•	.151	troy	ounces	gold/ton		

The "raw tailings"--(no crushing) were concentrated by gravity with a Mark VII Reichert Spiral and the concentrate assayed. The results ar

Head Assay(Troy Ounces Gold/ton	<u>c</u>	ONCENTRATE	CONCENTRATION RATI
0.319		5.37	16.83 : 1
0.043	. •	0.54	12.54 : 1
0.025	•	0.176	7.04 : 1

THIS PROVES THE ECONOMICS OF GRINDING-GRAVITY CONCENTRATION AS AN ECONOMIC METHOD TO RECOVER GOLD FROM THE APACHE JUNCTION TAILINGS.

SIGNED:

Richard F. Hewlett

DATED 7 March 17,198.5

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H-170-191	1/03/84	717.00	3904.07	33.54	12.75	10.19	3.65
H-170-192		348.00				20.00	
	1/13/04		3921.24		13.13		. 3.68
H-170-194	1/19/84	315.00	4340.68	~ 86.52	13.63	23.50	. 3.70
H-170-196	1/26/84	461.50	5240.32	77.35	17.85	21.70	. 5.01
H-170-197	1/31/84	866.50				35.'03	15.18
			. 16226.16	117.87	51.08		
TOTAL -	JANUARY	2728.00	33634.47	77.74	108.43	22.08	31.22
TOTAL -	YTD	2728.00	33634.47	77.74	108.43	22.08	31.22
M-170-200	2/03/84	. 520.00.	10153.87	119.93	31.18	22.82	7.32
M-170-201	2/13/84	1135.00	17981.37	109.92	62.3B	31.55	17.90
H-170-202	2/13/84	. 637.50			5.77	5.83	1.86
			1739.02	18.04			
H-170-203	2/17/84	488.00	7043.65	88.87	21.68	24.34	5.94
H-170-204	2/17/84	468,00	1307.15	18.30	4.28	12.06	2.82
H-170-206	2/21/84	288.50	908.14	20.23	2.92	17.60.	. 2.54
M-170-207	2/21/84	637.50	1202-80	15.16	4.83	• 4.68	· 1.49
H-170-200	2/28/84	. 291.20	2340.44	40.54	7.33	12.52	2.26
	FEPRUARY	4538.00	44979.68	53.87	140.37	18.05	44.14
TOTAL -	YTD	.7266.00	78614.15	45.81	248.81	20.07	75.36
H-170-211	. 3/15/84	413.50	2442.32	37.43	7.74	12.72	2.63
M-170-213	3/19/84	862.00	5331.67	37.15	16.73	10.43	· 4.60
H-170-214	3/22/84	1041.50	9127.17		29.30	14.92	7.77
H-170-216	3/26/84	680.50	13451.43	120.06 .	40.85	22.08	11.26
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H-170-217	3/26/84		4353.14	51.03	14.05	13.99	3.85
H-170-220	3/30/84	1158.30	17988.24	94.04	54.47	25.44	14.74
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H-170-222	4/03/84	475.00	4740.14	87.77	20.85	24.32	5.78
	4/03/84						1.83
H-170-223		· 622.00	1772.67	18.73	5.83	5.07	
H-170-225	4/06/84	390.75.	5401.65	85.54	16.71	25.55	4.99
H-170-226	4/12/84	507.00	7235.38	. 87.58	22.29	27.21	6.92
H-170-227	4/12/84	518.50	1347.30	17.01	4.41	4.92	1.79
H-170-228	4/16/84		4726.91	68.48	15.54	17.82	4.50
M-170-229	4/16/84	228.20	1107.40	21.54	3.65	7.85	1.33
H-170-231	4/23/84	1478.00	5600.99	24.72	18.27	8.03	5.93
H-170-234	4/30/84	751.50	2219.77	17.45 ,	. 7.38	4.58	2.47
TOTAL -		2337.25	36154.41	47.89	114.93	14.67	35.55
TOTAL -	TD	17512.75	167462.73	56.78	527.08	14.78	125.75
H-170-237	5/04/84	712.50	1390.08.	10.11	4.61	3.08	1.41
M-170-239	5/07/84	710.00	1987.09	17.85	6.34	4.93	1.72
M-170-241	5/07/84	251.00	2100.09	25.58	7.05	8.02	2.21
H-170-242	5/15/84						
		414.50	1266.06	20.36	4.24	6.01	1.22
M-170-244	5/24/84	967.50	2613.87	17.78	8.70	6.30	3.05
TOTAL - P	HAY	3237.50	7257.18	18.38	30.93	5.67	9.64
TOTAL - 1		21070.25					
			818181848	37.68	338.01	11.32	145.41
	The second second second			• .			
H-170-247	6/01/84	651.50	1686.93	16.44	5.34	5,18	1.45
H-170-248	6/06/84	370.50	608.27	10.48	1.74	3.31	0.61
H-170-250	6/11/84	400.00	1727.48	28,23	5.45	Q.16	1.42
H-170-252	6/15/84	. 370.50	1374.39	23. 71	4.67	7.01	1.37
H-170-254	6/25/84	949.50	3498.75 .	26.41	12.54	8.17	3.86
TOTAL - J							
		2762.00	9096.04	21.09	20.12	6.37	7.18
TOTAL - Y	פֿדי	23832.23	182815.75	29.39	288.14	8.84	174.59

"WASP" HIGH-GRADE VEIN ORE; IMPACTED-TABLE CONCENTRATED.

THE ABOVE ARE TYPICAL CONCENTRATE SALES.

••





Deochemical Laboratory, In

P.O. Box 1428 * Hesperia, California 92345 * Phone (619) 244-3481

Certificate of Analysis

66 CLIENT: DICK HEWLETT ATTENTION:

DATE: 03/12/85 CLIENT ORDER NO. INVOICE NO.: R20 LAB NO. : 1340 CC:

V.

ANALYTICAL METHODS: Au - Fire Analyses (1 A.T.) Ag - Fire Analyses

	SAMPLE #	Fire Au(oz/ton)	Fire Ag(oz/ton)
•	AJT AG DON'S TAILS	.014 .319	.05 .35
	AP JCT BUCKET AP JAC TAILS AP JCT BUCKET	.025	.50 .05 .20

off

Brett P. Crinelly Laboratory Manager

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(1)

J. B. LABORATORY

Specialists in Precious Metal Recovery

Pill

2702 S. 45TH ST. PHOENIX AZ 85034 (602) 966-8103

PROJECT PP											
SAMPLE #	D	ATE	•	PROC TO F	CESS RUN		WT TO USE	CON WT	DOR'E WT	ORILL WT	VOL ML
#1	3/1	L3/6	W2	-5 Sample	•		и (1) 1			_^ * =	1 T
ELEMENT	• .	·PI	PM	OZ PER TON HD ORE	OZ PER TON		ER TON		VA	WES	2014 g 14 1020
Au .				39.0			5 	•			
Ag				14.3					20 20 20		
			-				1 E			•	
	-				1. S.						
				•		?	•				
•									- 3 g = 1		
							-			•	
COMMENTS THANK YOU!											
the autho	ir i	7700			repared fo d above. J ublication for the p	.D. Ld	corat	OTY T	ACOT	nimer.	



Specialists in Precious Metal Recovery

2702 S. 45TH ST. PHOENIX AZ 85034 (802) 986-8103

PRO IECT DICK HEWLITT

	PP										
	SAMPLE #	DATE		PROC TO R	ESS IUN		WT TO USE	CON WT	DOR'E WT	ORILL WT	VOL ML
	#1 #2 #3	3/13/6	1	B21-41) ,						•	
	ELEMENT	P	РМ .	OZ PER TON HD ORE	OZ PER TON		ER TON		VAI	UES	
#1	Au .		.8					1			
	Ag			. 4							
								\top			
‡2 `	Au			.03				1			-
	Ag			.07							\neg
				· .			:	<u> </u>			\neg
3	Au			9.4							-
	Ag			5.5		•			-		\neg
	COMMENTS			- - -			7.	Kanje	. You]_	
				eport is provided of the specified of th			distr:	ibuti	on is		

쁖

J. B. LABORATORY

Specialists in Precious Metal Recovery

2702 S. 45TH ST. PHOENIX, AZ 85034 (602) 966-8103

SAMPLE #	GOLD	SILVER
	2.3	3.3 .
W2-3 1/16		2.0
Wasp Con	• 9 [°]	
W2 - 1 1/4	4.6	4.8
W2-2 1/4	3.8	4.5
W2-5 1/16	Repeat	
W2-4 1/4	1.2	2.0
	3.1	2.9
W2-5 1/4		1.9
W2-1 1/16	1.2	3.4
W2-2 1/16	2.8	
W2-4 1/16	1.2	2.7
LSL CON .	. 4	.7
Sil #12 15' 1/4	.2	.5
Wasp Red 1/16	.9	1.8
	2.4	4.1
Wasp Red		4.4
W2-3 1/4	2.1	
Wasp Red 1/4	1.9	2.7

1. 1 m

The above samples were submitted to us for fire tests to de-. termin values that may be refined from this material.

> Thank You: J.B. LAB

BOB HOLLADAY

P.13

Arizona Testing Laboratories

817 West Madison Street = Phoenix, Arizona 85007 = 602/254-6181

For: Hewlett Mineral Management Mr. Dick Hewlett 4304 West Mercury Way Chandler, Arizona 85224

Date: February 21, 1986

Lab. No.: 6225

Sample: Ore

Marked: See Below

Received: 2-18-86

Submitted by: Same

REPORT OF LABORATORY TESTS

GEOCHEMICAL ANALYSIS

		4
Samples Marked	Gold, ppm T. Og/Ton Silver, ppm	T. Og/Ton
W1 W2 W3 W4 V5 W6 Cons. W1 Cons. W2 Cons. W3 Cons. W4 Cons. W5 Cons. W6 Cons. W7 Cons. W8 Cons. W9 Cons. T Cons. G Ivan 1 Mag Light Mag Non Mag Ivan 2 Mag Light Mag Non Mag Ivan 3 Mag Light Mag Non Mag Ivan 4 Mag Light Mag Non Mag	$\begin{array}{cccccccccccccccccccccccccccccccccccc$	$\begin{array}{c} 0.44\\ \hline 2.19\\ \hline 0.26\\ \hline 2.23\\ \hline 0.26\\ \hline 2.23\\ \hline 0.24\\ \hline 0.06\\ \hline 1.84\\ \hline 0.99\\ \hline 0.93\\ \hline 0.99\\ \hline 0.99\\ \hline 0.93\\ \hline 0.93\\ \hline 0.93\\ \hline 0.93\\ \hline 0.03\\ \hline 0.06\\ \hline$
< = less than	Respectfully submitte	

Respectfully submitted,

ARIZONA TESTING LABORATORIES

CORRECTED REPORT March 13, 1986

Arizona Testing Laboratories

817 West Madison Street 🖵 Phoenix, Arizona 85007 🗆 602/254-6181

For: Hewlett Mineral Management Mr. Dick Hewlett 4304 W. Mercury Way Chandler, Arizona 85224

Date: March 13, 1986

Lab. No.: 6504

Sample: Ore

Received: 3/7/86

Marked: See Below

Submitted by: Mr. Shane Lohse

REPORT OF LABORATORY TESTS

GEOCHEMICAL ANALYSIS

Samples Marked: Gold. ppm T.og/T. Silver, ppm T.Oz/ Ton Carbon no prep 2.6 Lead no prep-1.7 320 9. WASP no prep 3 3 280 Yellow Aster-1/16 comp 680 8 80 2.5 no prep 1 Yellow Aster-1/16 comp 14 108 ₩ 2-1-1/16 < Channel Cut 1 370 n2-2 . 7' wide over 40 660 2-3 70: length 20 90 1000 2-4 74 530 Fit Bottom 2-5 60 .. 880 WASP Red-1/16 110 440 Silurian Table Cons .30 39 Sil #1 5. 1 225 13º Cons 25 5 50 458 17: 1.6 57 Sil #2 5' 24 15 42 10. 1 31 901 12. 17 7.3 Sil #3 5' 21 1 85 479 10' 12 27 12' 16 0.4: 45 Sil #4 5' 31 6.8 14 7 408 10' 0 1.2 23 <u>0.0</u> 12' 59 27 5.3 Sil #12 0 155 5' 7.9 10 292 10' 0. 2.7 08 30 15' Cons 874 9.3 14

Respectfully submitted,

6

ARIZONA TESTING LABORATORIES



HEWLETT MINERAL MANAGEMENT

ţ,



ne (cu	2) 961-0087 961-0072				4304 W. Mercury Way Chandler, AZ 85224
G(*:	OLDFIELD MI	NING DISTR	ICT-ARIZONA		• ·
	APITAL IVESTMENT	INVESTMNT INTEREST	INTEREST RETURN	CLAIMS ORE RESERV	ES NET OP. PROFI
\$	250,000	10 \$	\$ 1.44mm	B.King/ 240,000 t M.H. #2 @0.30	
\$	500,000	13 %	\$ 2.81mm	Indians 480,000 to @0.25	
\$	750,000	15 \$	\$ 5.40mm	Above + 720,000 to B.(Hand/Nite)	ons\$ 36,000,00
\$	1,000,000	18 😫	\$ <u>8.64</u> mm	T.Thumb & 800,000 to Mammoth 2/3 @0.30 A	ons \$ 48,000,00
\$	2,000,000	28 %	\$ 26.88mm	B.Queen & 1,600,000 to M.Hubbard & @0.30 A	
\$	3,000,000	37 \$	\$ 66.69mm	Hammoth(Pat), Annex, 6 2,400,000 to Laurence @0.35 A	
\$ 8	r,000,000	49 %	\$194.04mm	All of the above: Plus options.	\$ 396,000,000
≡=: Gol Na	etter ette etter etter	\$300/ troy o	unce	Plus options.	y 376,000,00

	by certify that the fo	191; Veno V6J	2W7	dward Aver B.C.	ne fest		STUS	ION SUPERIN ENDER ST. V 254-1647 T CERT	TENDENCE C VANCOUVER ELEX 04-5075	PATORIE OMPANY (CANDA) L B.C. CANADA VEA T IA CABLE SUPERVI E OF ASSA
		GOLD	SILVER	1	T	Gro	• •		•	
a <u>a transporta</u>	MARKED	oz/st	02/8t		'xcxt				. 2012 ·	· 555
	•				2 					
	30401 30402 30403	0.102 0.335 7.846	0.18 1.20 2.92	C LUS D. C LUS D. C D NC EX		+20 -20 50m SI		TARLE	OCT. 10	137
	30404 30405 30406	0.022 0.148 13.120	0.50 0.49 5.14	STOCK	SIAN DE PILE FAL ENTRATE	MILL	FEED	а	T. DEE E Det	
5	30407 3040 <u>3</u> 3040 3	0.100 0.012 0.016	0.19	15- FT.	RIT IN PIT IN	SAMPLE	IN TRE	New S.	WE SHA DOTH C. ERTICAL	E SHAFT
	30410	0.005	_ .	<u> </u>	E ALVAEL	• <i>Sdut</i>	DA ESO	TRENON	1994 1994 1994 1995 1997 1997 1997 1997 1997 1997	
						₽				
									-	-
TS CON	TS RETAINED ONE MONTH S AND REJECTS WILL BE S S ARE THE CONFIDENTIAL CLUSION OR EXTRACTS FR YOUT OUR WRITTEN APPI THE FEE CHARGED	PROPERTY OF OM OR REGAR	CLIENTS PU CLIENTS PU DING OUR R LIABILITY AT	OF ONE YEAR.		Specialists	Surveyors	, Inspecto		TAL ASSAYER Prs. Weighers

MEAIBER American Society For Testing Materials - The American Oil Chemists' Society - Canadian Testing Association REFEREE AND OR OFFICIAL CHEAISTS FOR National Institute Of Oilseed Products - The American Oil Chemists' Society

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CHAS. A. DIEHL

Phoenix, Arizona, Nov 6 29

zona Assay DILINI

and particular

Phone 4447

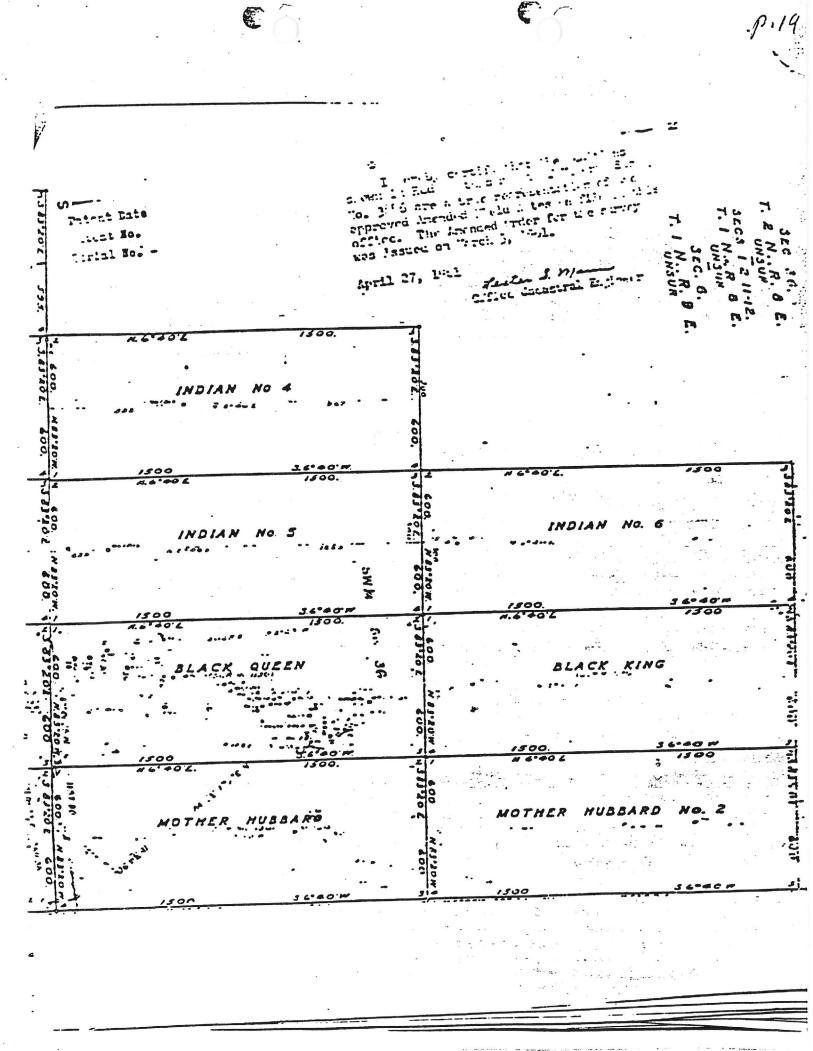
315 North First Street

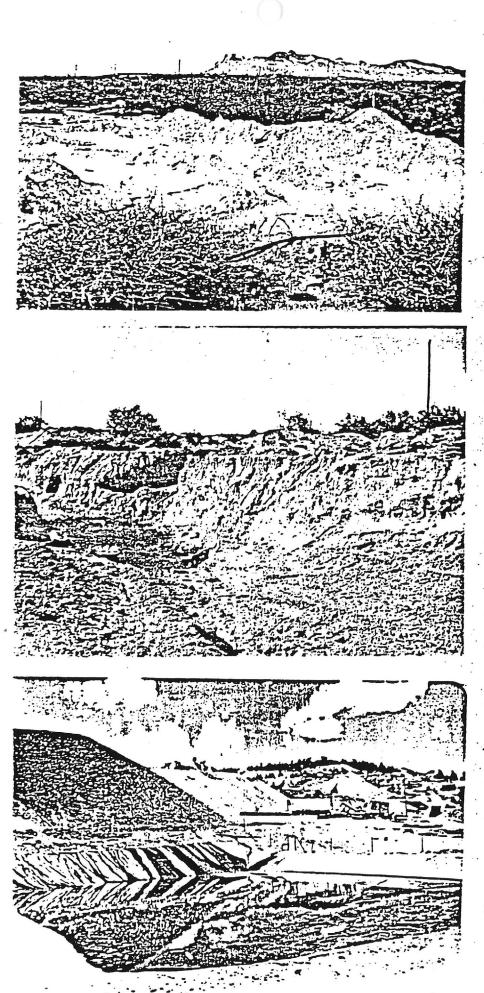
P.O. Box 1148

Elis Critifies That the samples submitted for assay by Apache Trail Gold MJ. Co.

contain as follows per ton of 2000 lbs. A voir .:

SAMPLE MARKED	SILVER	VALUE AT	GOLD OUNCES PER TON	VALUE AT	PE+		PERCENTAGE		
		UUG PEROL	OUNCES PER TON	S20 PER OZ.				!	
"Black Lucen" 150' level						•			
5" width 14' N: of cross cut.	.5	\$. 25		\$2.80					
2 19" width 10½" N. of cross cut.	1.3	\$.65					· · · · ·		
5 Country rock hang- ing wall.	.7 -	\$.35		2					
Footwall Oxidized Granite	.7	\$.35	1. 14	- \$21.60					
5 7'-0" N. of cross _cut-7之" wide.	1.2		12.84						
) Quartz & Altered nite 3' N. cross	ut.		.02	\$25.20 \$.40	ان المراجع المراجع				
18" Honging Wall Country Rock 3'-0"" North of crosscut.	1.0	\$.50	.02						
10" Country Rock in Hanging & Footwall 75-0". N. of crosscut	. 1.0	ş. 50	.03	\$. 60					
Special	19,1	ទ្ធំ9.55	43 - 59	\$871.60	: -			-	
	4 • • •							-	
		ي يومون ال						-	
								•	
								: * .	
Charges 5 11.25 Jaid			Assayer	has	A, c	Die	hl.	• • •	
						• • • • • • • • • • • • • • • • • • • •			





T. INGS: Beginning in T8>, the "high-grade" vein ore was mined underground (down to 1,000 feet). The ore was crushed by "STAMP MILLS" and the coarse gold was trapped by mercury tables. The tailings were discharged with precious metals still un-recovered.

The grade of the tailings ranges from 0.05 troy ounces gold/ton on the surface to v 0.319 troy ounces gold/ton at depth(18 feet).

THE FACING TABLE PRESENTS THE MONTHLY NET PROFIT FOR GRADES OF 0.05 to 0.20 AND PRODUCTION RATES OF 10-120 TPHT

<u>HIGH-GRADE VEIN</u>: A highgrade vein has been mined within the last year with a hydraulic back-hoe. The gold will be concentrated using the same plant as used for the tailings. This will result in a higher-grade product.

The grade previously mined from the vein was 4 troy ounces/ton, but lower-grade portions of the vein can also economically be mined.

THE FACING TABLE PRESENTS THE MONTHLY NET PROFIT FOR GRADES OF 0.25 to 4.00 AND PRODUCTION RATES OF 2-20 TPHT.

HEAP LEACHING: Lower-grade portions of the vein system will be excavated and hauled to a pad for "heap leaching". The ore will be leached by a solvent and the precious metals recovered by zinc ppt.

Considerable ore exists from 0.10 to 0.50 troy ounces gold/ton. THE FACING TABLE PRESENTS THE MONTHLY NET PROFIT FOR GRADES OF 0.04 to 0.15 AND FOR PRODUCTION RATES OF FROM 10,000 to 150,000 TONS/MONTH.



HEWLETT MINERAL MANAGEMENT

-.... F. HEWLETT Phone (619) 242-8277 242-5737 15110 Mandan Apple Valley California 92307

p.21

GOLDFIELD, ARIZONA POTENTIAL ORE RESERVES

My feeling from the beginning was that this could be consolidated into a \$ Billion District. Study of the underground data and analysis of surface veins previously been mined and drilling confirms a large potential.

VENTURE	ORE TYPE	TONNAGE	GRADE (A	VALUE(\$ <u>GROSS</u>	300 Au)(1 NET
Apache Mining J.V.	•	135,000.	.215	\$8.707	\$7.358
	0.PWasp	3,000,000.	.30	270.	240.
-	Nettie	1,000,000	.50	150.	140.
	Hilltops	3,000,000	.30	270.	240.
Leases	Total	4,000,000	.10	120.	80.
GOLDFIELD MINES/ Darrel Hand	Tailings	200,000	.215	12.9	· 10.9
Court Case Under advisement	Open-Pit	10,000,000	.25		650.
for a decision by Apeals Court Judge.			.« 	_	368.258

(P	2	7	~	

TA

GOLDFIELD DISTRIC LASH FLOW BY PROCESSING .

10,000

HOD:

90,000

1,200,000

840,000

480,000

120,000

ILINGS;	MONTHLY NET	OPERATING E	PROFIT!	
.20 Au/T	\$ 100,000	\$ 300,000	\$ 600,000	\$ 900,000
.15 Au/T	70,000	210,000	420,000	630,000
.10 Au/T	40,000	120,000	240,000	360,000

30,000

.05 Au/T

10 Tons/Hr. 30 Tons/Hr. 60 Tons/Hr. 90 Tons/Hr. 120 Tons/Hr PRODUCTION RATE/HOUR:: FOR 10 HOURS/DAY(20 DAYS/Month)

60,000

OPEN-PIT HIGH-GRADE VEIN: MONTHLY NET OPERATING PROFIT!

4 Au/T	\$ 472,000	1,180,000	2,360,000	3,540,000	4,720,000
2 Au/T	\$ 232,000	580,000	1,160,000	1,740,000	2,320,000
l Au/T	\$ 112,000	280,000	560,000	840,000	1,120,000
.5 Au/T	\$ 52,000	130,000	260,000	390,000	520,000
.25 Au/T	\$ 22,000	55,000	110,000	165,000	220,000

10 Tons/Hr. 15 Tons/Hr. 20 Tons/Hr. 2 Tons/Hr. 5 Tons/Hr. PRODUCTION RATE/HOUR:: FOR 10 HOURS/DAY(20 DAYS/Month)

OPEN-PIT HEAP LEACHING: MONTHLY NET OPERATING PROFIT!

.15 Au/T	\$ 370,000	\$ 925,000	1,850,000	3,700,000	5,550,000
.10 Au/T	\$ 220,000	\$ 550,000	1,100,000	2,200,000	3,300,000
.08 Au/T	\$ 160,000	\$ 400,000	\$ 800,000	1,600,000	2,400,000
.06 Au/T	\$ 100,000	\$ 250,000 .	\$ 500,000	1,000,000	1,500,000
.04 Au/T	\$ 40,000	\$ 100,000	\$ 200,000	\$ 400,000	\$ 600,000
~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~~					
	10,000	25,000	50,000	100,000	150,000
	PRODUCTION	RATE PER MON	TH-UNDER LEAG	CH	

COSTS: \$ 8/tgn for heap Leaching

\$ 10/ton for processing tailings

\$ 20/ton for mining/concentrating open-pit high-grade vein.

265 S. EXTENSION

-----

6/06/84

6/11/84

6/15/84

6/25/84

-170-250

-170-252

-170-254

TOTAL - JUNE

TOTAL - YTD

370.50

400.00

390.50

949.50

23832.25 185815.95

2762.00

608.27

1727.68

1374.39

3698.75

9096.04

10.48

28:23

23.91

26.41

21.09

29.39

1.94

5.65

4.67

12.54

30.15

588.16

3.31

8.16

7.01

8.17

6.37

8.84

MESA, ARIZONA 852L

JUL 13 1984 -

CONTENT

T.O./LOT

3.65

3.68

3.70

5.01

15.18

31.22

31.22

9.32

17.90

1.86

5.94

2.82

2.54

1.49

. 2.26

44.14

75.36

2.63

4.60

7.77

3.85

11.26

44.84

5.78

1.83

4.99

6.92

1.79

4.50

1.33

5.93

2.47

35.55

1.41

1.75

2.21

1.25

3.05

9.66

1.69

0.61

1.63

1.37

3.85

9.1E

174.59

165.41

155.75

120.20

. 14.74

... ...

AF HCAE MINING & DEVELOPMENT - 1984 CONCENTRATE ACCOUNT ECONC GOLD >>> < < < SILVER > > > < < < LOG DATE NET TOTAL ASSAY CONTENT ASSAY NUMBER RECEIVED WEIGHT PAID T.O./TON T.O./LOT T.D./TON -170-191 1/03/84 717.00 3906.07 35.56 12.75 10.17 -170-192 1/13/84 368.00 3921.24 71.37 13.13 20.00 -170-194 1/19/84 315.00 4340.68 86.52 13.63 23.50 -170-196 1/26/84 461.50 5240.32 77.35 17.85 21.70 -170-197 1/31/84 866.50 16226.16 51.08 117.89 35.03 TOTAL - JANUARY 2728.00 33634.47 77.74 108.43 22.08 TOTAL - YTD 2728.00 33634.47 77.74 108.43 22.08 -170-200 2/03/84 520.00 10153.89 119.93 31.18 35.83 -170 - 2012/13/84 1135.00 19981.59 109.92 62.38 31.55 -170 - 2025.77 2/13/84 639.50 1739.02 18.04 5.83 -170-203 2/17/84 488.00 7043.65 88.87 21.68 24.34 -170-204 2/17/84 468.00 1309.15 18.30 4.28 12.06 -170-206 2/21/84 288.50 908.14 2.92 20.23 17.60 -170-207 1503.80 2/21/84 637.50 15.16 4.83 4.68 -170-208 2/28/84 361.50 2340.44 40.54 7.33 12.52 TOTAL - FEBRUARY 4538.00 44979.68 53.87 140.37 18.05 TOTAL - YTD 7266.00 78614.15 65.81 248.81 20.07 -170-211 413.50 2442.32 3/15/84 37.43 7.74 12.72 -170-213 3/19/84 845.00 5331.67 39.15 16.93 10.63 -170-214 3/22/84 1041.50 9127.17 56.27 29.30 14.92 -170-216 680.50 3/26/84 13451.63 120.06 . . 40.85 33.08 -170-217 13.99 3/26/84 550.50 4353.14 51.03 14.05 -170-220 17988.24 54.47 3/30/84 1158.50 94.04 25.44 TOTAL - MARCH 4709.50 52694.17 66.33 163.34 18.46 TOTAL - YTD 11975.50 131308.32 66.07 412.15 19.27 475.00 -170-222 4/03/84 6740.14 87.77 20.85 24.32 -170-223 4/03/84 622.00 1772.67 18.73 5.83 5.87 -170-225 4/06/84 390.75 5401.65 85.54 25.55 16.71 -170-226 4/12/84 507.00 7235.38 87.5B 22.29 27.21 -170-227 4/12/84 518.50 1347.30 17.01 4.41 6.92 -170-228 454.00 4/16/84 4726.91 68.48 15.54 19.82 -170-229 338.50 4/16/84 1109.60 21.56 3.65 7.85 1478.00 -170-231 4/23/84 5600.99 24.72 18.27 8.03 -170-234 4/30/84 751.50 2219.77 19.65 7.38 6.58 TOTAL - APRIL 5537.25 36154.41 47.89 114.93 14.69 TOTAL - YTD 17512.75 167462.73 56.98 527.08 16.98 -170-237 5/04/84 912.50 1390.08 10.11 4.61 3.08 -170-239 5/07/84 710.00 1887.08 17.85 6.34 4.93 -170-241 5/07/84 551.00 2100.09 25.58 7.05 . 8.02 -170-242 5/15/84 416.50 1266.06 20.36 4.24 6.01 -170 - 2445/24/84 967.50 2613.87 17.98 8.70 6.30 TOTAL - MAY 3557.50 9257.18 18.38 30.93 5.67 TOTAL - YTD 21070.25 176719.91 37.68 558.01 11.32 -170-247 6/01/84 651.50 1686.93 16.44 5.36 5.18 -170-248

p.2	1.	•		HETAL REF	IKERS, LTD				
1.2	7 .	APACHE	KINING & D	E ENT	- 1983 CO	CENTRATE	ACCOUNT (C	(מיזאכ	
					< < < 60	LD >>>	< < < SIL	VER >>>	
	LOS	DATE	KET	TOTAL	ASSAY	CONTENT	ASSAY	CONTENT	
	NUKBER	RECEIVED	WEIGHT	PAID	T.D./TON	T.O./LOT	T.O./TOX	T.D./LOT	
	BALANCE FO	IRWARD	162384.99	205228.21	58.98	618.35	34.83	369.64	
	H-170-179	12/07/E3	453.00	8418.40	102.57	24.78	31.29	7.56	
	E-170-180	12/07/E3	341.00	872.41	16.00	2.73	5.00	0.85	
	H-170-152	12/12/83	860.00	4916.57	36.53	15.71	10.75	4.62	
	H-170-184	12/14/83	712.50	3345.91	30.18	10.75	8.03	2.86	
	R-170-185	12/19/83	941.00	5972.06	42.31	19.91	10.69	-5.03	
	H-170-189	12/22/83	353.00	3761.10	70.30	12.41	21.73	3.84	
	K-170-190	12/28/83	508.50	2981.31	38.33	9.75	10.78	2.79	
	DECEMBER T	DTAL	4199.00	30267.76	48.03	96.04	14.07	27.55	
	YTD TOTAL		166583.99	235495.97	53.51	714.39	24.45	397.19	

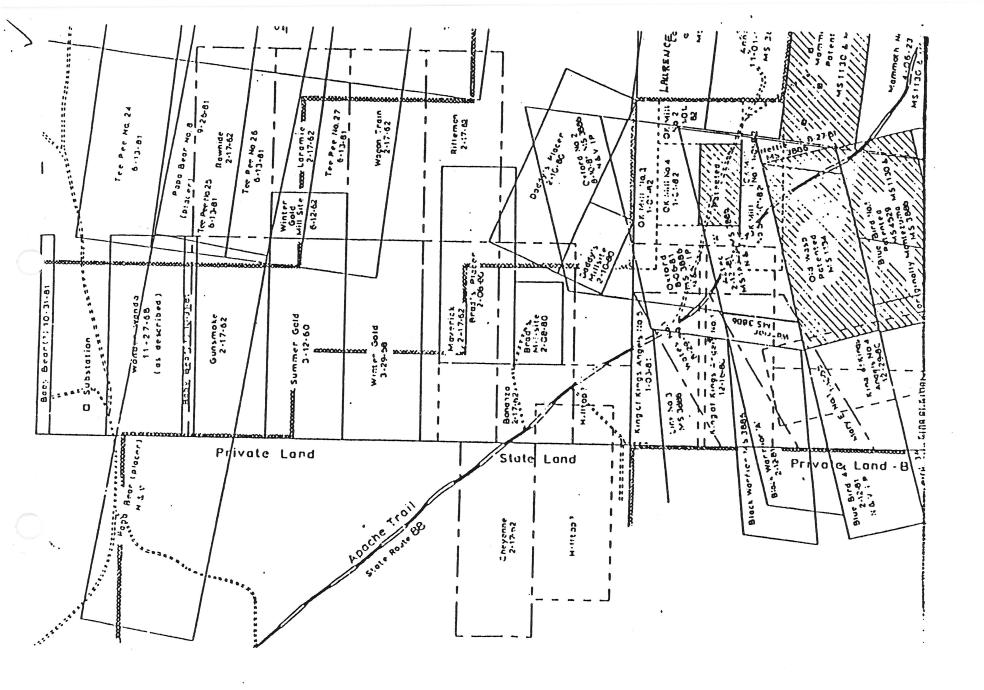
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· . • .				<u></u>			P * * *	id. to tay	-1.4 -1
•	-	~		1			201	5. E.	CAOR MES
,					•		I.L	SA Fin	IA 25202
								16021 554	-2704
711	2/84	APACHE	KINING & DE	ETAL REFIN	- 1983 CO)		CCOUXT		•
111.	2/84	NINGAL I			( ( 601	D >>>	< < < SIL	ER >>:	
1.0	-	DATE	NET	TOTAL			ASSAY	CONTENT	
LD		RECEIVED	WEIGHT	PAID		T 0 /10T	T.D./TON		
	EER		215.00	3307.20	90.95	9.78	24.85	2.67	
H-170-		6/27/83 KOT PURCH.	0.00	0.00			Contraction of the second		
				3307.20				2005 Carteria	
JUKE				3307.20			24.85		
YTD TI	ILAL		215.00	2201.10	10.17	3.10	£1100	2.0.	
					101.74	17.95	27.86	4.92	
K-170		7/18/83	353.00	6112.61	125.89				
H-170-		7/19/83		8438.21					
K-170		7/22/83		11682.13					
K-170		7/26/83		6770.B1					
JULY				33003.76					
YTD T	OTAL		2047.50	36310.96	98.35	106.43	27.23	21.10	
							10.15		
K-170	-120	B/01/83		6477.50				5.51	
K-170	-131	8/05/83		3376.54					· .
H-170	-132	\$/10/B3			44.63				
- H-170	-133	8/15/83			113.60				
H-170	-134	B/19/83	309.75	10066.52	192.90				
, H-170		8/22/83	322.00	6153.17	112.32	18.08		5.10	
K-170		8/19/83	1735.00		SECONDS		SECONDS		
H-170			519.00	10527.53	120.61	31.30	34.01	8.83	
. M-170		8/30/B3		8202.75			33.16	6.86	
K-170		8/26/83	50000.00		SECONDS		SECONDS		
AUSUS				64348.33		190.96	30.83	62.47	
YTD T				100659.29				92.25	
	UIML		2/013/00	14403.112.					
M 194		9/06/83	477 00	10704.36	78.61	31.41	63.04	20.08	
H-170		9/12/83		5890.39					
H-170				3662.74					
H-170		9/16/83							
		9/20/83		4690.98					
		9/21/83		2642.81		25.09		16.04	•
		9/23/B3		8445.98					
		9/27/83	1176.00			6.39			
		8/30/83	365.00			100 million from the			
		TOTAL	4936.00						
YTD T	DTAL		62779.00	147305.90		477.01		•••••	
						11.14	18.62	7.14	
			766.50	3242-14	29.08			3.40	
		10/11/B3	625.50						
		10/14/83	970.50					5.42	•
		10/21/83	BOB.00						
		10/24/83	630.50					. 7.33	
		10/27/83	739.50			5000 at 198			
		10/21/83	503.50				SECONDS		
		10/31/83	71000.00		SECONDS	00 (5		40.51	
OCTOB		ITAL	96044.00	2//94.52	39.01				
T GTY	OTAL		158823.00	175100.42	60.63	522.24			
		121 12043-44				10 19	11 7	12.66	
		A 11/09/B3	572.75						
		11/09/83		1860.78					
H-170	-173	11/14/83		1550.94				B 1.54	•
		11/14/93		3297.24					
		11/18/23		6826.57					
		11/18/83	705.50	1888.12	17.83				
NOVER		TOTAL	3561.99	30127.79	57.33				
YTD T	DTAL		162384.99	205228.21	58.98	618.35	34.83	369.64	
					-				

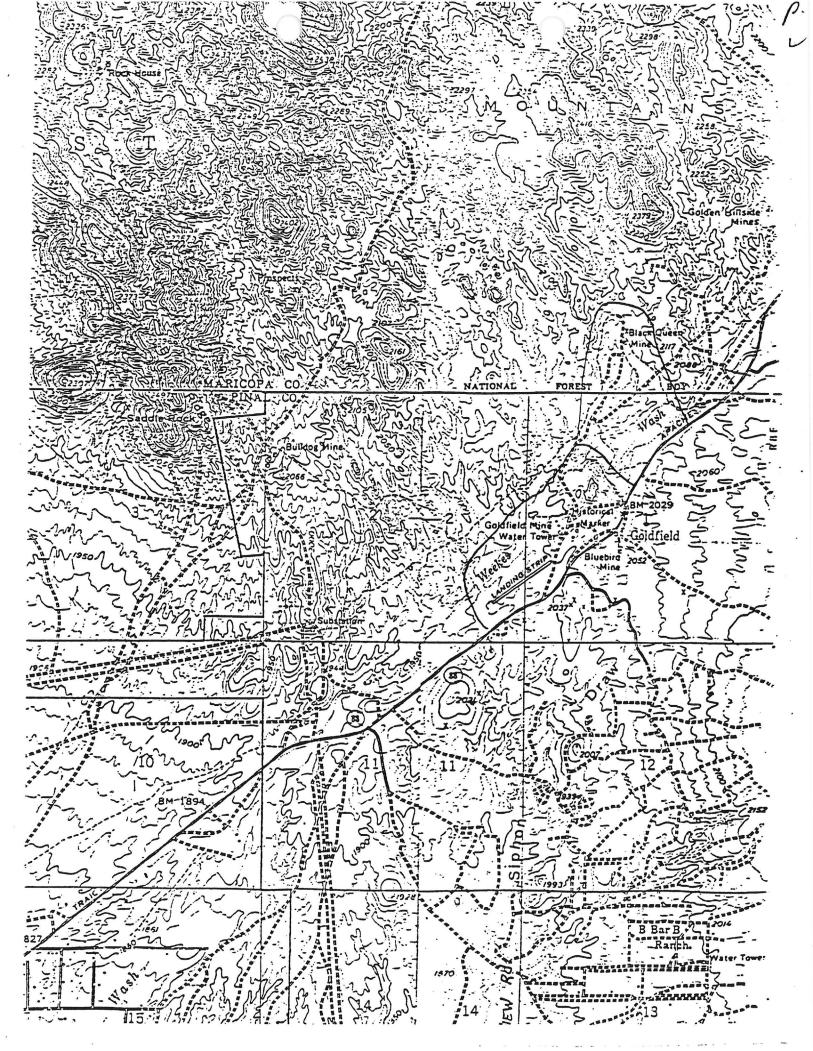
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eochemical Laboratory, Inc

P.O. Box 1428 * Hesperia, California 92345 * Phone (619) 244-3481

## Certificate of Analysis

66 CLIENT: MR. D. HEWLETT ATTENTION:

DATE: 03/18/85 CLIENT ORDER NO.: INVOICE NO.: R209( LAB NO. : 1350 CC:

ANALYTICAL METHODS:

Au - Fire Analyses (1 A.T.) Ag - Fire Analyses

9

SAMFLE #	Fire Au(oz/ton)	Fire Ag(oz/ton)
CIN-50	trace	.25
CIN-100	.011	.45
CIN-200	.003	.35
D-50	.349	.65
D-100	.245	.60
D-200	.517	.45
JIM'S TAILS	.003	.25
JIM'S CRUSHED	trace	.25
FRED'S HEADS	trace	.25

Q

Frett P. Connelly Laboratory Manager

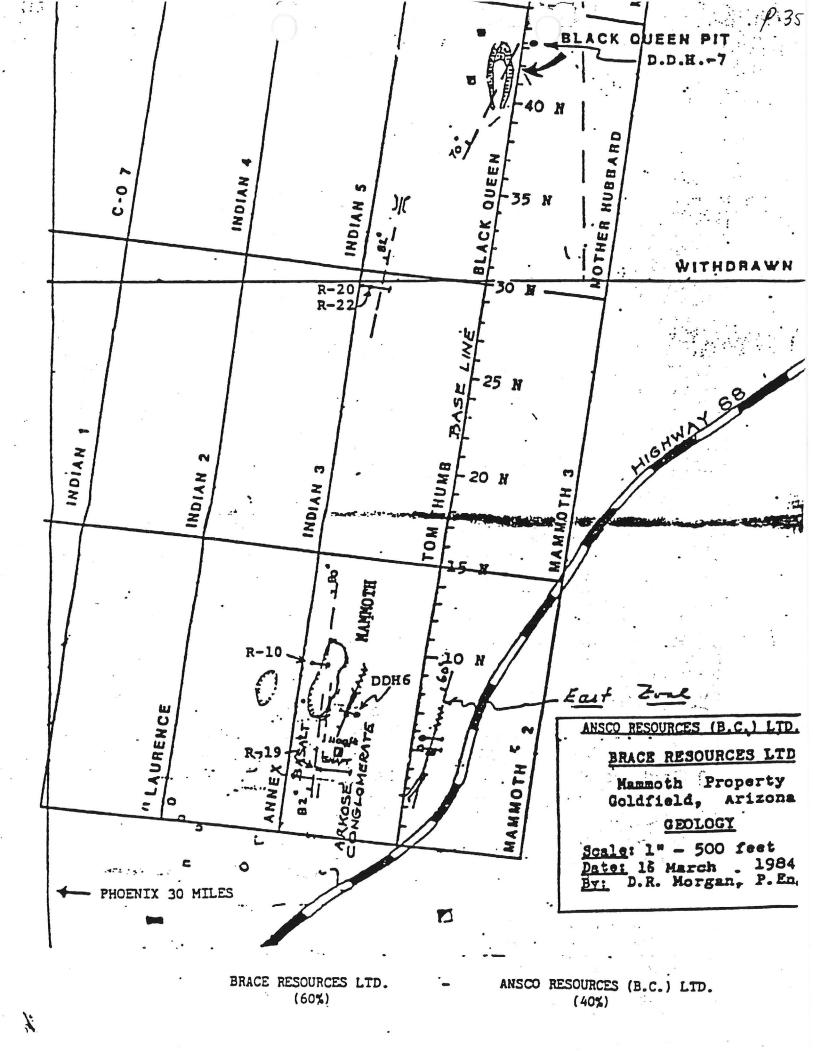
### BLACK QUEEN MINE

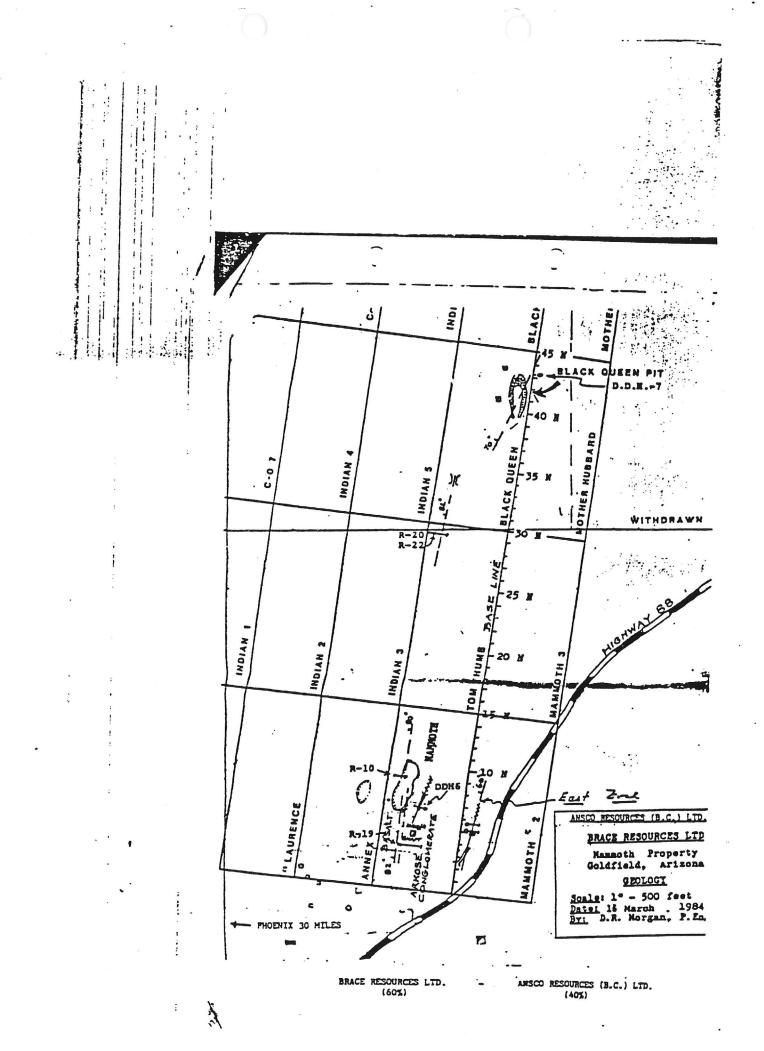
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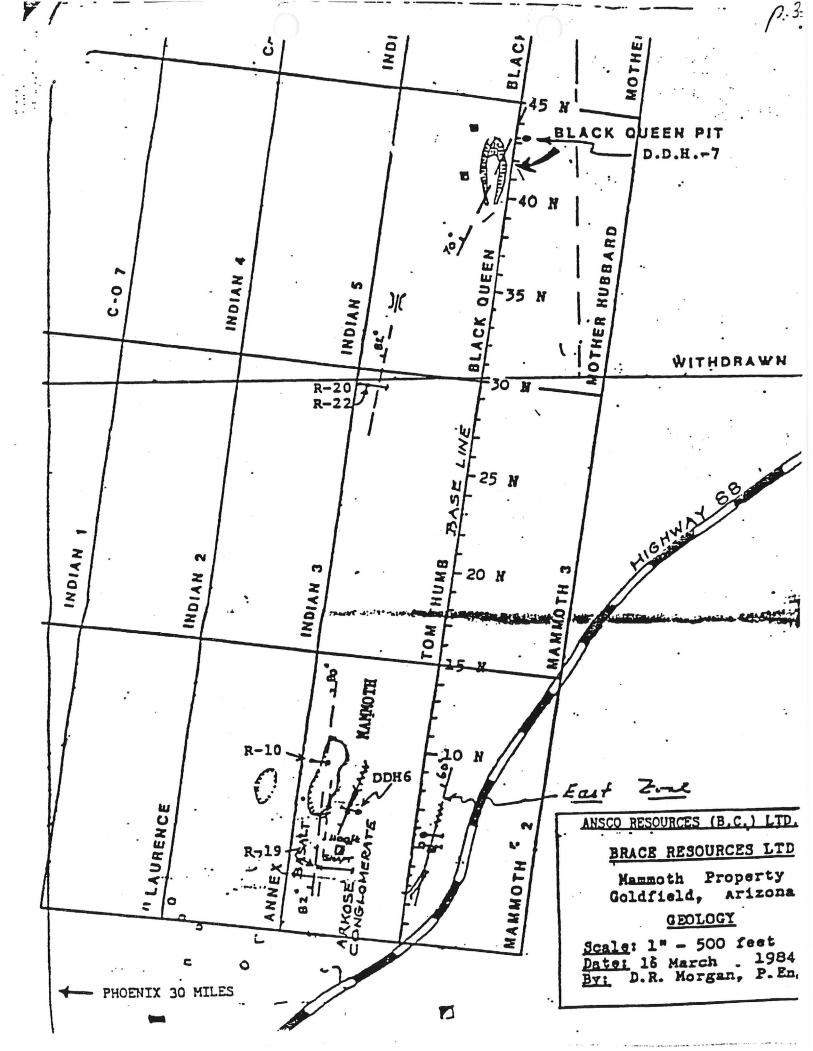
### MARICOPA COUNTY

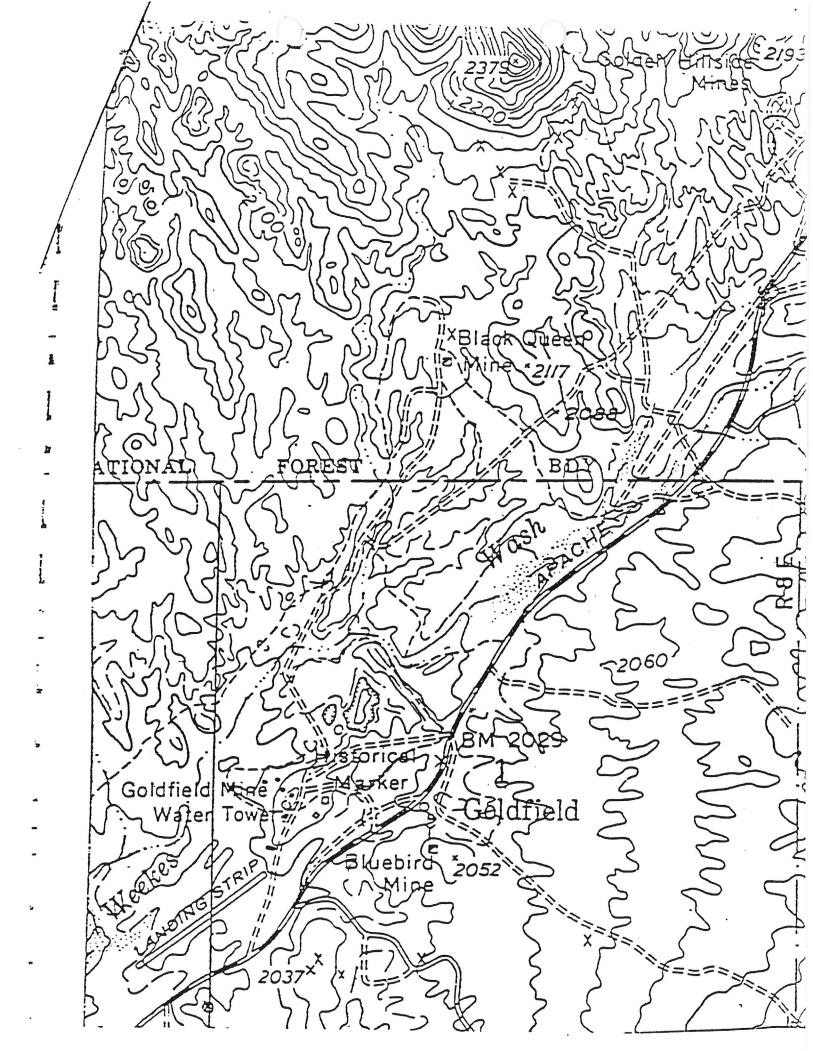
RRB WR 2/15/85: Visted the Mammoth Mine, Goldfield District. Jerry Sandstead reports that the Black Queen is still tied up in litigation which is delaying development of the property.

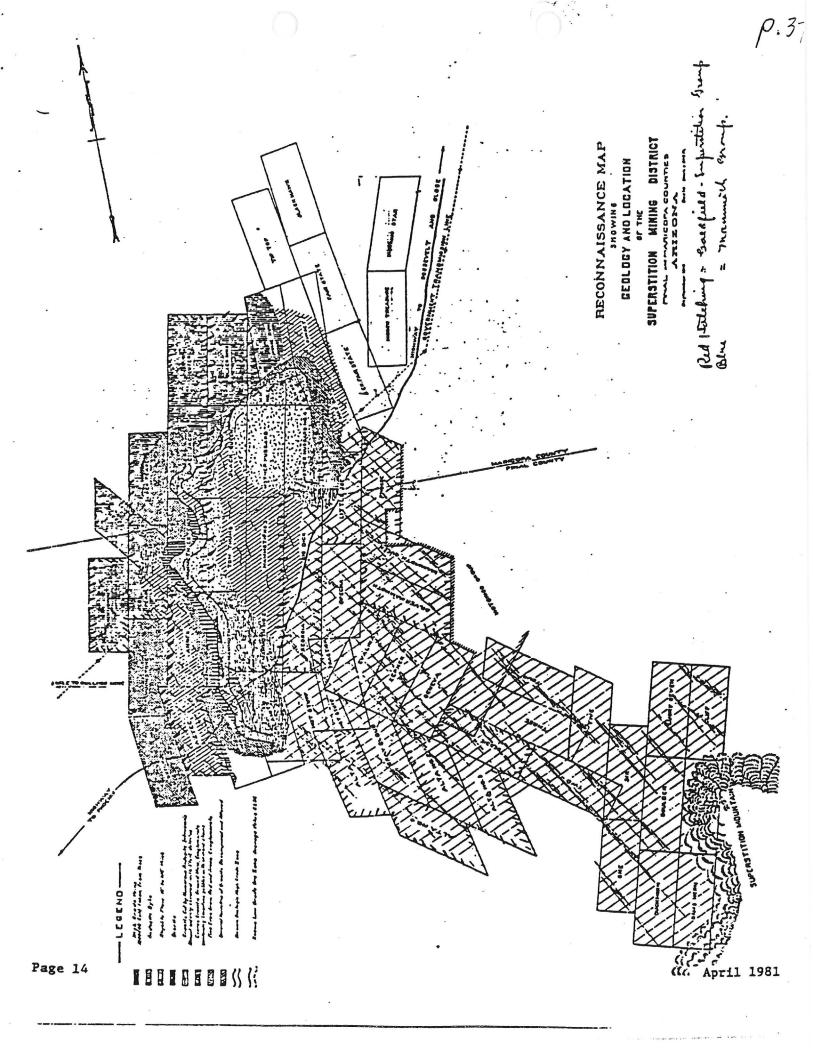
NJN WR 1/10/86: Michael Sheridan PhD geology professor at Arizona State University called seeking information on mining activities in the Goldfield Mining district. Mr. Sheridan believes there is a group shipping from the wedge south of the Mammoth Group (Black Queen Mine - file) to the Inspiration Smelter. He is mapping in the area to lead a field trip on the Arizona Geological Society sponsored spring 1986 symposium. He speculated that the Gold in the district may be remobilized from placer in the areas red beds by Tertiary volcanic driven hydrothermal circulation into faults and reactive rock units. Although he observes only small narrow veins and pocket deposits in the area he believes there is potential for a large deposit.

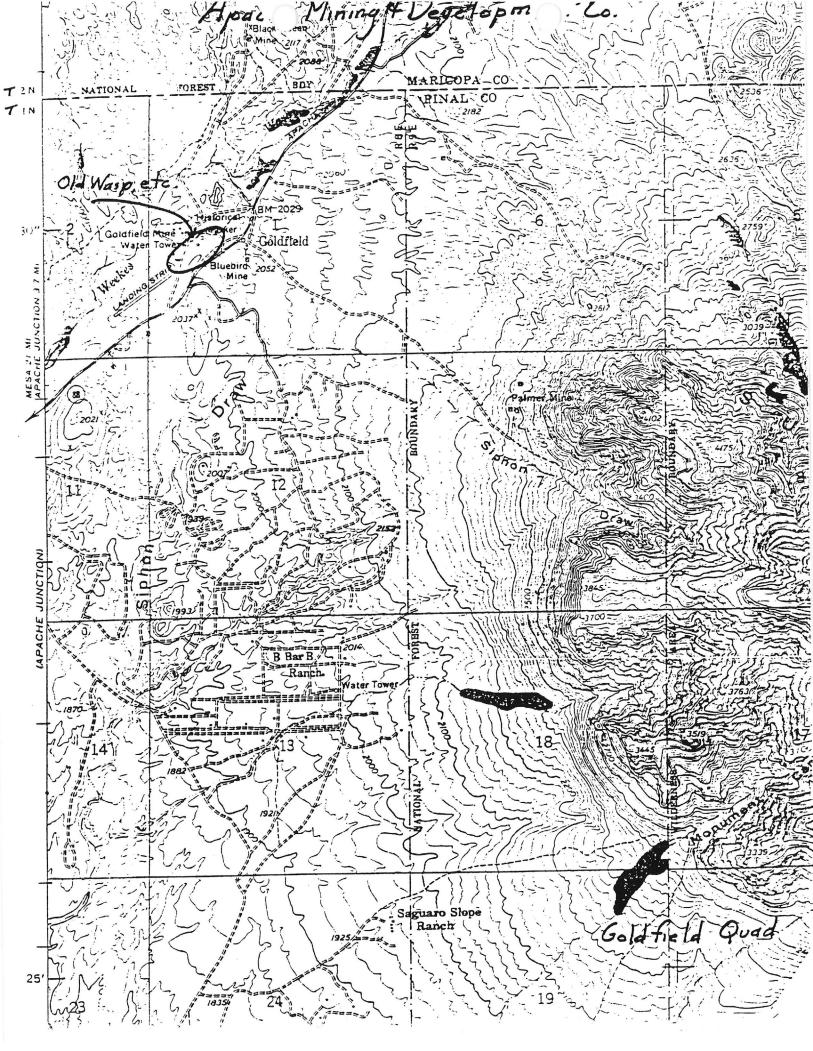


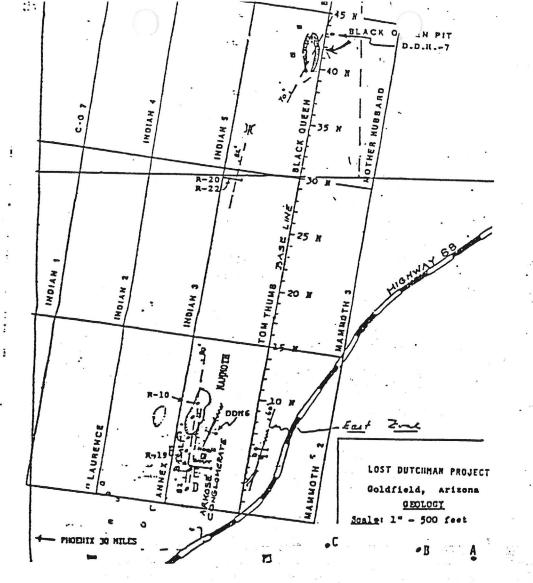












DL:... PROJECT SAMPLING

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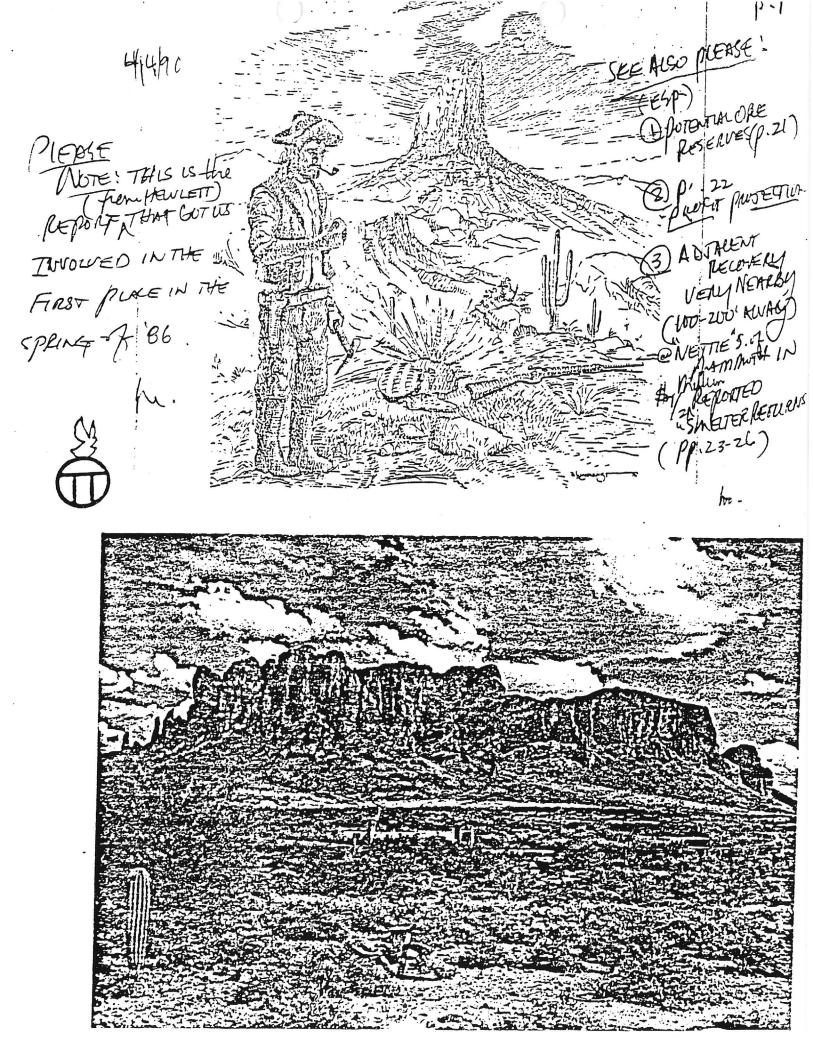
SAMPLE	TYPE	SAMPLER OF LABORATORY	Troy Ounces/Ton GOLD SILVER
Α	Surface Outcrop	J.B. Laboratory	
8	Surface Outcrop	J.B. Laboratory	
c	Surface Outcrop		
0	Surface Outcrop	J.B. Laboratory	.20 1.10
Ε	Surface Outcrop	Root & Norton	.110 0.30
Tagin Sina F	Surface Vein	Root & Norton	.466 1.12
6	Surface Vein	Brace Resources	.88
M		Brace Resources.	
	Surface Vein	Brace Resources	19.00
64 R	Rotary Drill-Hole	Brace Resources 0-200' depth	.082
	· · · · ·		
77 R	Rotary Drill-Hole	B.R.: 0-100' depth	.188
	Rotary Drill-Hole Diamond Drill-Hole	B.R.: 0-100' depth	.188
)DH 6	Rotary Drill-Hole Diamond Drill-Hole Diamond D.Hole	B.R.: 0-100' depth B.R.: 0-100' depth	.188 .142 .129
DDH 6 IDH 7	Diamond Drill-Hole	B.R.: 0-100' depth B.R.: 0-100' depth Brace:84'-86.5'	.188 .142 .129 .154 7.21
00H 6 10H 7 1-10	Diamond Drill-Hole	B.R.: 0-100' depth B.R.: 0-100' depth	.188 .142 .129 .154 7.21 .512 .45
P7 R DDH 6 200H 7 1-10 1-19 1-20	Diamond Drill-Hole Diamond D.Hole Rotary D.Hole	B.R.: 0-100' depth B.R.: 0-100' depth Brace:84'-86.5' B.R.:165'-180'	.188 .142 .129 .154 7.21

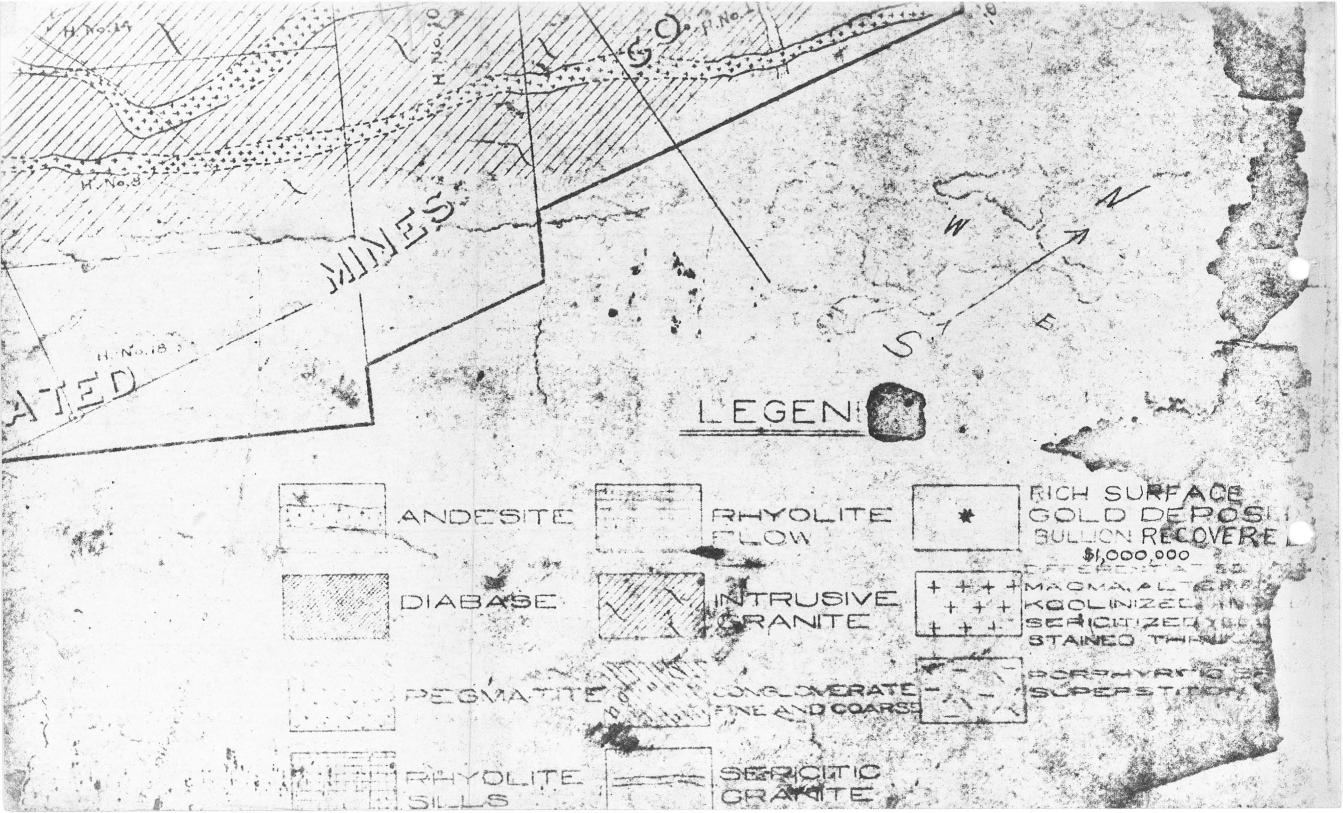
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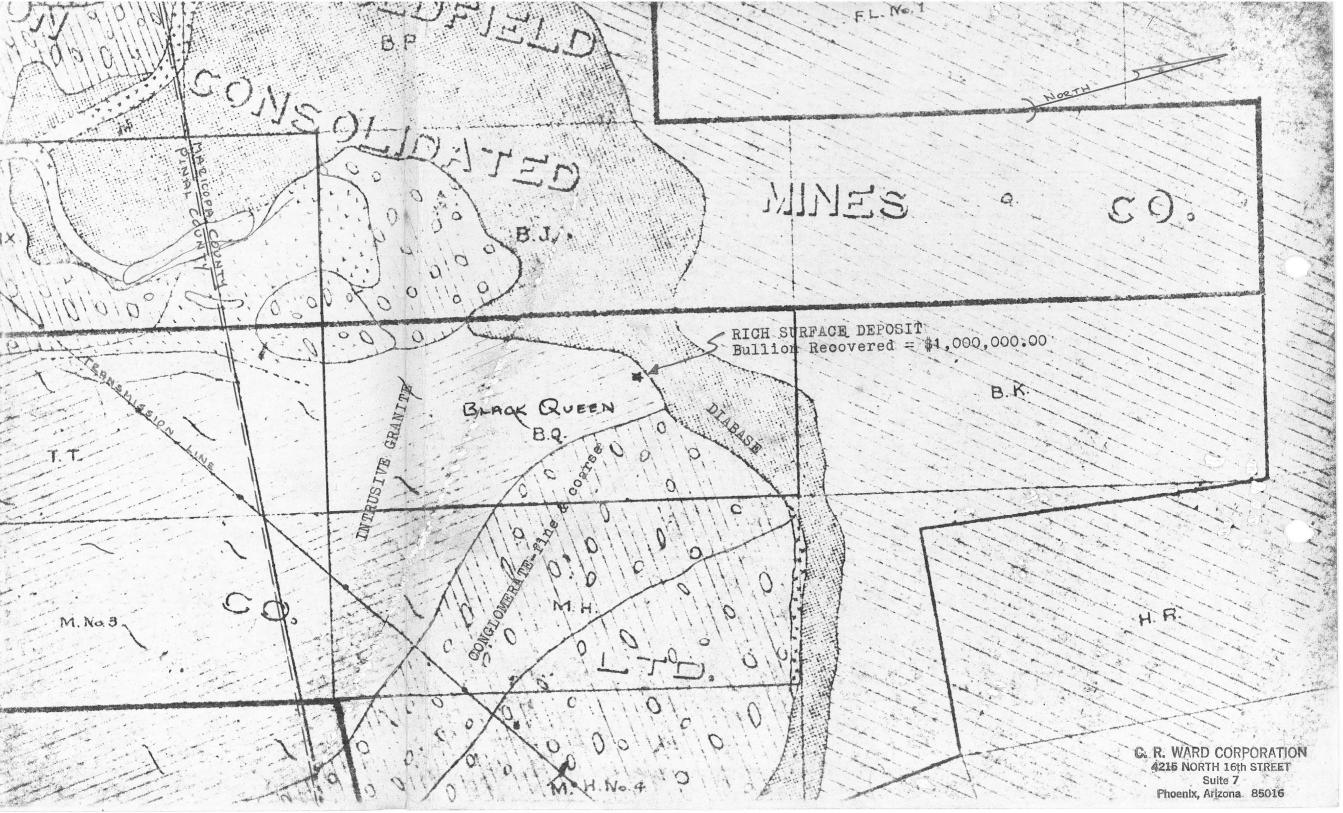
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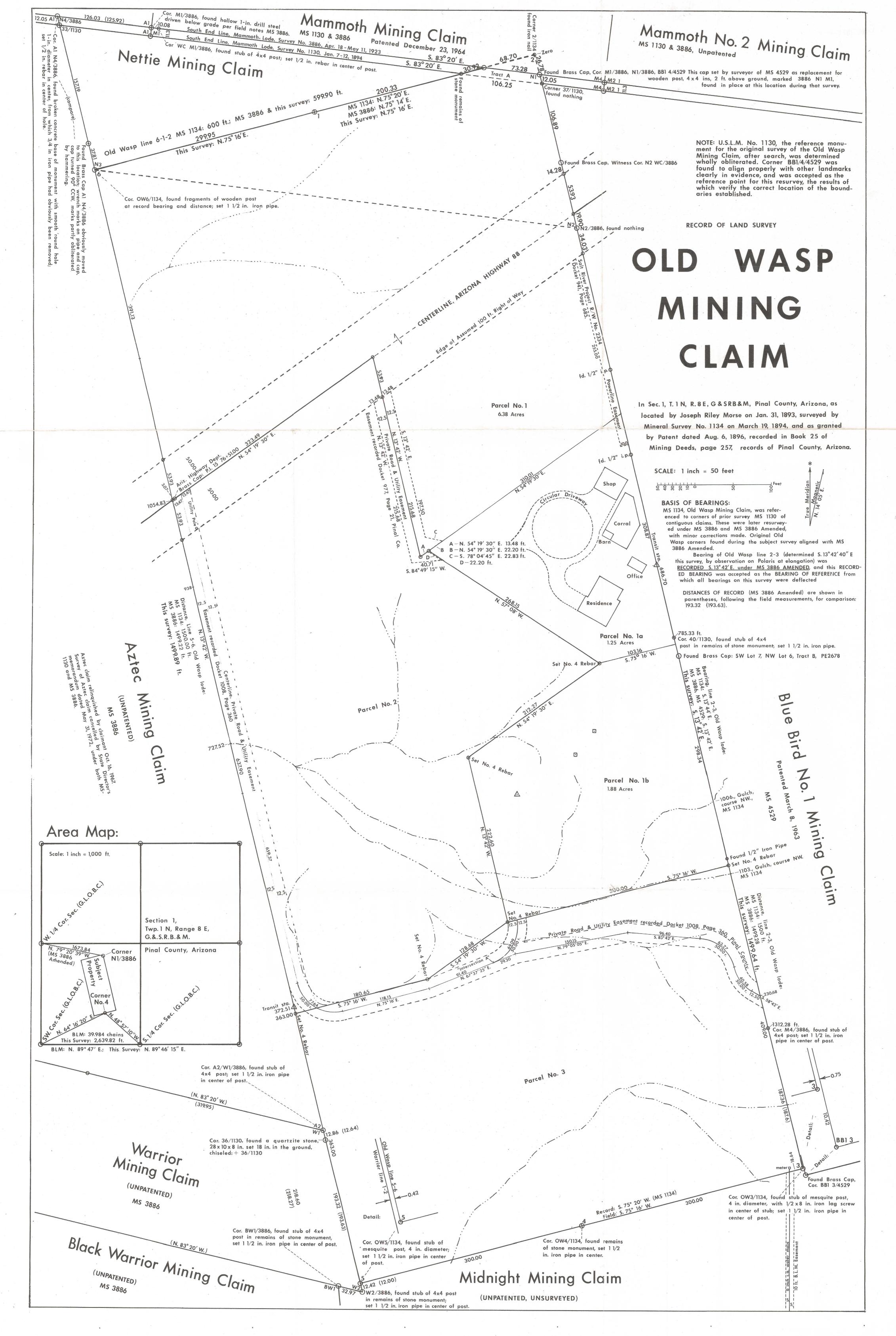
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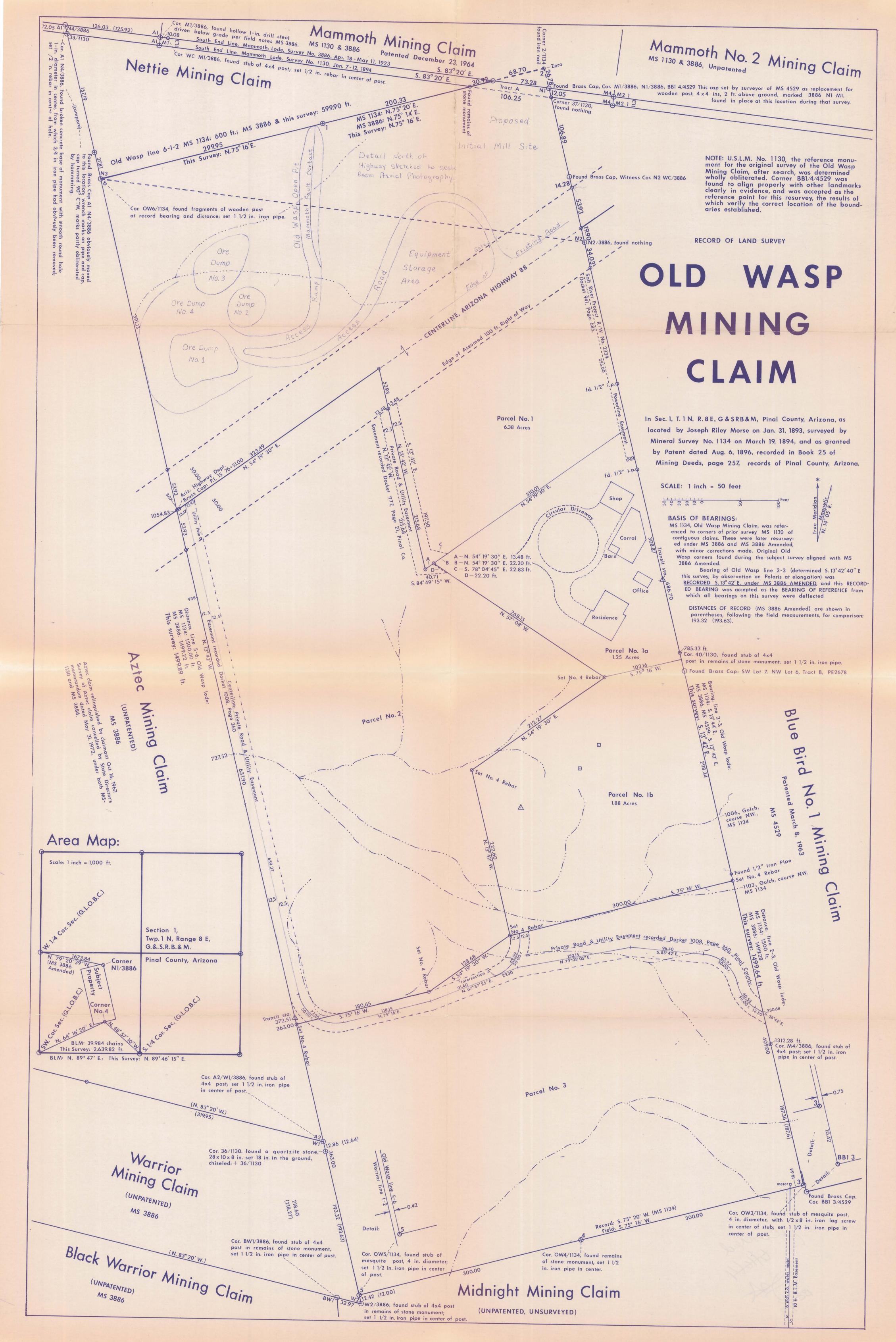
underground and surface = 0.191 troy ownces/ton head ore.











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